

INDUSTRIAL APPLICATION OF FLOTATION  
MODELS

PART I

by

LARRY ARTHUR CRAMER



Submitted in partial fulfilment of the requirements of the  
degree of Doctor of Philosophy in the Department of Chemical  
Engineering, University of Natal.

DURBAN,  
South Africa.  
January, 1975.

### ACKNOWLEDGEMENTS

I would like to especially express my thanks to Professor E.T. Woodburn, my supervisor, for his guiding hand and infinite patience. Also I thank him for allowing me to publish the mathematical method for circuit analysis by the gamma function model which he developed.

I am also indebted to the Johannesburg Consolidated Investments Corporation for the opportunity to extend my education in a practical environment.

TABLE OF CONTENTS

	<u>Page</u>
I. List of Figures	5
II. List of Tables	7
III. The Rustenburg Deposit and its Treatment	
A. Formation of the Deposit	8
B. Mineralogy of the Deposit	9
C. The Relationship between Values and Structure	9
D. Sulphide Minerals and Their Association	11
E. Treatment of the Rustenburg Deposit	12
IV. Present Flotation Modelling Status	17
V. Fundamentals of Flotation and Characteristics of the Models	20
A. The Flotation Process	22
B. Structure of the Chemical Analogy Models	24
C. Possible Limitations in the Model Structure	25
VI. Application of the Four Flotation Models	36
A. Measurement of the Residence Time Distribution	37
B. Performance of the Single Phase Models	42
1. The Single Rate Model	42
2. The Rate plus Non-floating Model	52
3. The Gamma Function Model	69
D. Performance of a Two-Phase Model	88
E. Conclusions about the Four Flotation Models	104
VII. Use of the Gamma Function Model as a Circuit Design Aid	115
A. Simulation of Various Hypothetical Circuits	116
B. Comparison of Actual Circuit Data with Simulations	121
C. Conclusions	122

	<u>Page</u>
VIII. Mathematical Formulation of the Models	128
A. The Single Rate Model	128
B. The Rate plus Non-floating Model	129
C. The Gamma Function Model	130
D. The Two-Phase Model	140
E. Parameter Estimation Techniques	141
IX. Test Methods and Data Collection	149
X. List of Symbols	155
XI. Bibliography	157
XII. Part Two - Raw Data and Programming	163



LIST OF FIGURES

	<u>Page</u>
1. Flotation of Various Ores	15
2. Typical Sulphide Conglomerate	16
3. Sulphide Recovery	30
4. Chalcopyrite Recovery	31
5. Pentlandite Recovery	32
6. Pyrrhotite Recovery	33
7. Effect of Acid Addition	34
8. Effect of Changes in the Quality of Water	35
9. Rougher Bank Residence Time Distribution	40
10. Cleaner Bank Residence Time Distribution	41
11. Batch Fits of the Single Rate Model	47
12.           "                   "	48
13.           "                   "	49
14. Continuous Fits of the Single Rate Model	50
15.           "                   "	51
16. Batch Fits of the Rate plus Non-floating Model	56
17.           "                   "	57
18.           "                   "	58
19. Continuous Fits of the Rate plus Non-floating Model	59
20.           "                   "	60
21. Batch Parameter Prediction of Continuous Operation for the Rate plus Non-floating Model	61
22.           "                   "	62
23.           "                   "	63
24. Predictions for Refloated Material - Rate plus Non- floating Model	66
25.           "                   "	67
26. Batch Fits of the Gamma Function Model	73
27.           "                   "	74
28.           "                   "	75
29. Gamma Function Rate Constant Distribution	76

	<u>Page</u>
30. Continuous Fits of the Gamma Function Model	77
31.       "                               "	78
32. Batch Parameter Prediction of Continuous Operation for the Gamma Function Model.	79
33.       "                               "	80
34.       "                               "	81
35. Predictions for Refloated Material - The Gamma Function Model	84
36.       "                               "	85
37. Importance of the Froth Phase	97
38.       "                               "	98
39. Batch Fits of the Two-Phase Model	99
40. Sum of Squares Surfaces for the Two-Phase Model	100
41. Batch Predictions by the Two-Phase Model	101
42. Batch Fits of the Two-Phase Model with a Non-floating Fraction	102
43. Batch Prediction by the Two-Phase Model with a Non- floating Fraction	103
44. Froth Removal Rate Batch Tests	107
45.       "                               "	108
46. Mean versus Froth Removal Rate	109
47. Variance versus Froth Removal Rate	110
48. Flotation Feed Batch Tests	117
49.       "                               "	118
50. Hypothetical Circuits	119
51. Simulated Grade-Recovery Curves	120
52. Tested Circuits	123
53. Flotation Cell	148
54. Continuous Operation Curves	154

LIST OF TABLES

	<u>Page</u>
I. Flotation Variables	21
II. Effective Flotation Cell Volumes	38
III. Simple First Order Rate Constants	44
IV. Batch Test Prediction of Continuous Operation - Rate plus Non-floating Model	64
V. Predictions for Refloated Material - Rate plus Non-floating Model	68
VI. Batch Test Prediction of Continuous Operation - Gamma Function Model	82
VII. Predictions for Refloated Material - Gamma Function Model	86
VIII. Predictions for Refloated Material - Gamma Function Model with a Non-floating Fraction	87
IX. Froth Removal Data - Batch Tests	93
X. Froth Removal Data - A Continuous Test	96
XI. Single Phase Model Parameters and Froth Removal Rate	111
XII. Model Comparison - Prediction of Continuous Operation	112
XIII. Model Comparison - Predictions for Refloated Material	113
XIV. Model Comparison - Batch Test Description	114
XV. Data Summary for Rougher-Scavenger Circuit	124
XVI. Data Summary for Circuit Analysis	126
XVII. Test Circuits - General Remarks	127



### III. THE RUSTENBURG DEPOSIT AND ITS TREATMENT

#### Formation of the Deposit

The Bushveld Igneous Complex is one of the world's most famous geological phenomena and consequently has been subject to a large amount of study since the 1900's. Formed over 2000 million years ago, the complex is of an oblong shape, its long axis stretching from somewhat east of Burgerfort, to Zeerust, RSA, in the west, a total length of 288 miles. On its north-south axis the complex begins just north of Pretoria and extends north for 153 miles.

The origin of the Bushveld Complex is still a subject of some debate; originally believed to be the result of a differentiation of an 18000' thick magma, the most popular theory today is that proposed by F.C. Truter in 1955.<sup>1</sup> Truter proposed a series of magmatic intrusions from five different points. This long continuous process of repeated intrusion is said to have given rise to the layered structure. Differentiation then plays a minor part in formation of the complex.

A second theory suggests the advent of a series of extensive sheets, each one being differentiated separately, thus the formation of regular layers without cross-cutting.<sup>2</sup> A relatively new theory which sounds far-fetched but is geologically supported, proposes the occurrence of three nearly simultaneous impacts from comet-like bodies which resulted in either the production of voluminous magma by fusion of crystal and mantle rocks or by a severe thinning of the earth's crust and later magmatic intrusion.<sup>3</sup>

Other theories have been proposed, including metasomatism (transformation) of the upper Transvaal sediments,<sup>4</sup> but none receives the support of the above three theories.



### Mineralogy of the Deposit.

Regardless of the origin of the Bushveld Igneous Complex, the structure and associated mineralogy has been well defined. E.R. Schmidt in 1948 made a good study of the Rustenburg area in his Master's Thesis.<sup>5</sup> In the Rustenburg area, the Merensky reef strikes eight degrees east of true north at a dip of eight to ten degrees. Along its plane the reef undulates and varies widely in width. Associated with the famous Merensky reef are several other mineral bearing bands, in descending order they are :

1. mottled anorthosite
2. hang-wall spotted gabbro
3. the Merensky reef
4. a small chrome band (not always present)
5. a 10 to 24" band of coarse felspathic pyroxenite
6. a large chrome band,  $\frac{3}{4}$ " or greater
7. a narrow band of pure anorthosite
8. spotted anorthosite gabbro

In normal mine terminology, groups 1, 2, 7 and 8 are referred to as norite, group 3 as Merensky reef and groups 4, 5 and 6 as reef.

At times, where the structural disturbances described and explained by Schmidt exist, some of the bands disappear momentarily. In dome structures the coarse felspathic pyroxenite band is absent near the top of the dome while in the circular depressions the larger chrome band is absent.

### The Relationship between Values and Structure.

The PGM (platinum group metals) and sulphide minerals appear to be in association except in the chrome bands where

the PGM value outweighs sulphide value. PGM value is concentrated in the chrome bands and overlaying coarse felspathic pyroxenites; however sometimes PGM value is found in the spotted anorthosite gabbro when the band is high in sulphides. The Merensky Reef, the hanging-wall spotted gabbro and mottled anorthosite always contain relatively very small amounts of PGM and sulphides.

It has been management policy in the past to separate "waste" and "reef" in times of a high market demand for PGM, and when the market falls to re-process the waste material. Thus for increased production the spotted or mottled rock (primary norite) is hand-sorted by unskilled labour such that the majority of plant feed consists of the chrome bands and the coarse felspathic pyroxenite. Unfortunately during the course of this investigation, various portions of the "waste" material were processed with the run of mine ore. Varying mineralogies have an effect on the flotation performance, as illustrated in fig. 1. Evaluation of plant performance of the various mineralogies, in terms of strict model parameters would be an expensive and time-consuming operation.

Several mineralogical investigations of the RPM (Rustenburg Platinum Mine) ores have been reported for the benefit of the Johannesburg Consolidated Investments.<sup>6</sup> These reports have been studied and main points of mineral association are discussed below. No reference is made with regard to PGM minerals and their associations for two reasons: 1) all modelling work references total sulphide or flotation of various sulphide minerals, and 2) the information concerning PGM mineralogy and production is considered confidential by the company and cannot be publicised.



### Sulphide Minerals and their Association.

Various sulphide minerals are present in the Rustenburg ore and are listed here in order of their relative abundance. The four major sulphide constituents are Pyrrhotite ( $\text{Fe}_7 \text{S}_8$ ), Pentlandite ( $(\text{Fe}, \text{Ni})_9 \text{S}_8$ ), Chalcopyrite ( $\text{Cu Fe S}_2$ ) and Pyrite ( $\text{Fe S}_2$ ). Minor sulphides are Mackinawite ( $\text{Fe S}$ ), Vallenite ( $\text{Cu}_3 \text{Fe}_4 \text{S}_7$ ), Cubanite ( $\text{Cu Fe}_2 \text{S}_3$ ), Bornite ( $5 \text{Cu}_2 \text{S Fe}_2 \text{S}_3$ ) and Bismuthimite ( $\text{Bi}_2 \text{S}_3$ ).

The mineralogical researchers agree that sulphides crystallized during the later stages of magmatic differentiation, with the pyrrhotite probably crystallizing first. As a consequence, sulphides appear primarily as massive aggregates which have an interstitial relationship with the silicate minerals (primarily pyroxene and plagioclase feldspars). Normally they are moulded against the feldspar and/or pyroxene crystals; in the case of pyroxene, the sulphides often form very fine grained inter-growths along cleavage planes. These sulphide patches vary in size from one millimeter to several centimeters, however they are always accompanied by a number of very small sulphide flecks attached to the feldspar and pyroxene crystals. Thus while the majority of sulphide is liberated from gangue material at an early comminution stage, a significant proportion of the sulphide remains attached to gangue materials even after fine comminution.

The sulphide aggregates themselves have a complex nature and generally consist of varying proportions of the major sulphide minerals. Separation by grinding, of the various mineral species, particularly separation of pentlandite and chalcopyrite from pyrrhotite would require extremely fine grinding. The aggregates tend to have a specific structural nature, in which pyrrhotite forms a core intermittantly covered by pentlandite and then by chalcopyrite, with pyrite crystals

often evident in the pyrrhotite and pentlandite. (See Fig. 2).

#### Treatment of the Rustenburg Deposit

The regularity of the strike and dip of the reef in the Rustenburg area has made mine planning and development a rather simple problem, the only interference having come from the circular depressions and the domes. Like gold mining, the valuable material is contained in very small "reef" thus stoping is done in rather narrow bands following the dip of the reef. The typical mine stoping width is 28 inches, this includes the chrome band, approximately 19 inches of the overlying pyroxenites and 9 inches of the underlaying norite. Large proportions of the norite are sorted on the stoping face and used as fill for supports. The ore is then transported by means of a number of haulage ways to the hoist where it is raised and loaded onto trains for transportation to the reduction works. The RPM reduction works may be broken into three sections, the modelling investigation applies to only one of these sections, the flotation section. The other sections, comminution and gravity concentration, may contribute to some of the problems encountered in the flotation modelling, and hence they are described briefly here.

From the mine the ore passes through jaw crushers reducing large lumps to minus six inch, from there the ore is passed over washing screens and the plus two inch is sorted for waste. The remaining coarse material is crushed in two stages to minus half inch for ball mill feed. The feed is stored in silos, where retention time will vary up to 60 hours depending upon production schedule. Fine comminution is done in two-stage ball milling, all mills operating in close circuit with hydro-cyclones.

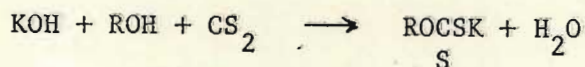
Following the milling section is the gravity concentration



section. In this section the heavy "metallic" PGM values are trapped in corduroy cloth. Since a large amount of labour is employed here, considerable attempts have been made to by-pass this section, in fact for the bulk of the test work the corduroy section was not in operation.

Flotation follows the regular practice of normal sulphide flotation, all the sulphides are floated since the main economic value lies in recovering PGM metals not in the production of pure copper and/or nickel concentrates. The flotation circuit is a simple one, normal operation consists of a roughing and a cleaning section. The circuit was changed for certain tests so that a flotation model could be used to predict plant performance in tune with circuit design, potentially, modeling's largest advantage.

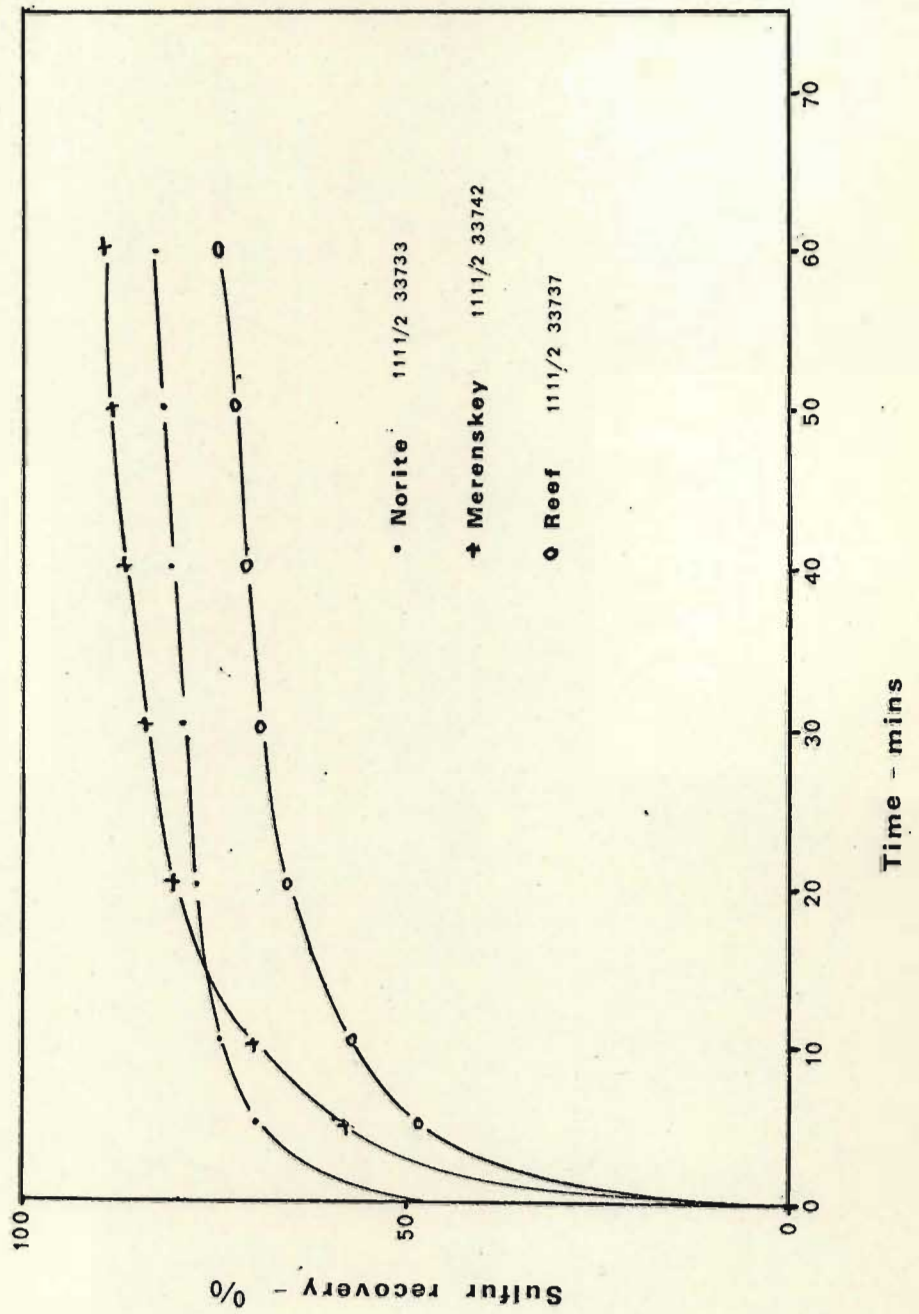
The reagents used in flotation are common to sulphide flotation. Xanthate is added as the collector, the mechanism of rendering the mineral surface hydrophobic is not completely understood, however it is assumed that the anionic portion of the molecule attaches to the mineral surface while the hydrophobic carbon-chain tails provide the water repellant surface. Xanthates are formed by reaction of alcohols with carbon disulphide, as follows :



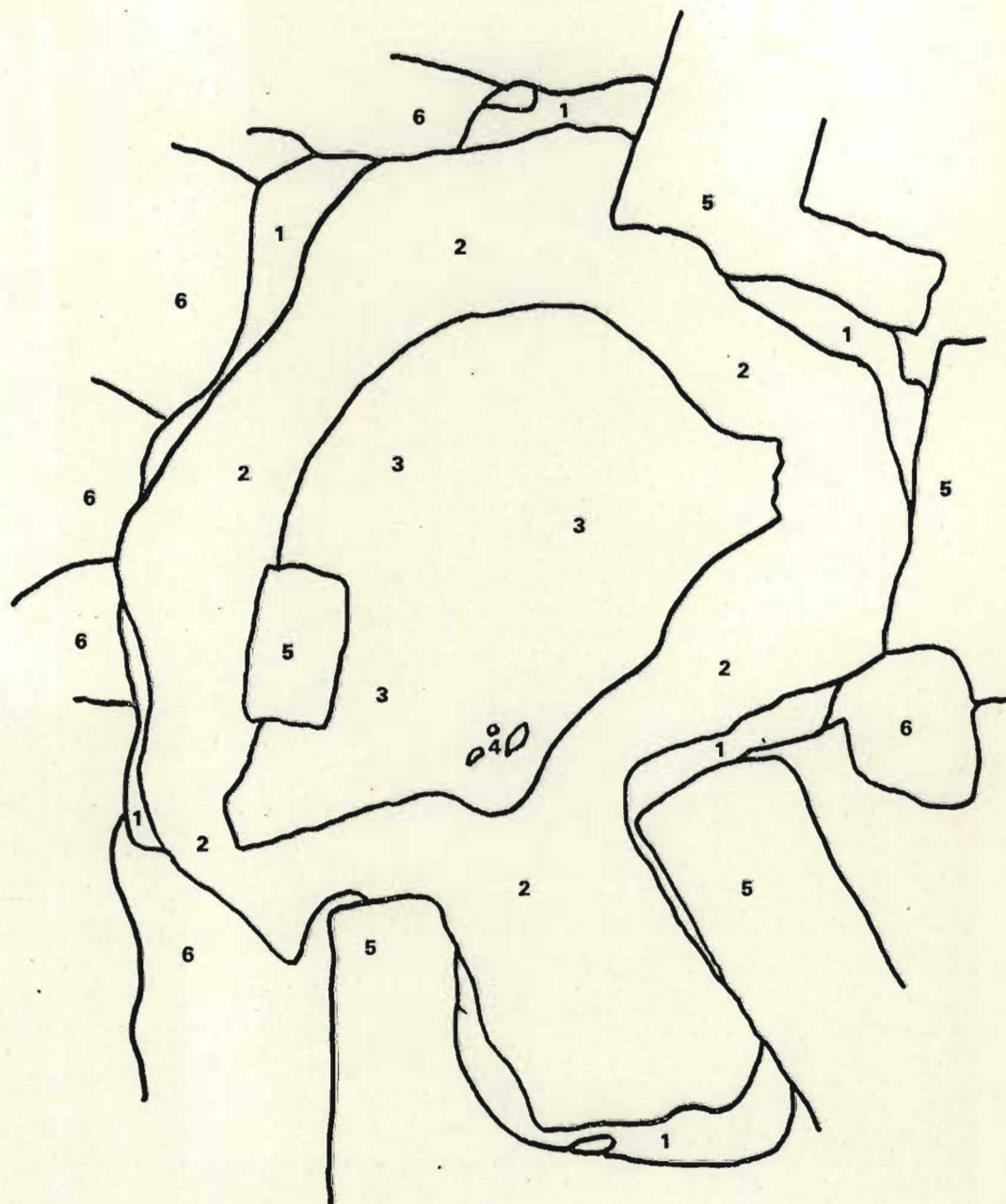
Copper sulphate is added as an activator, primarily for the pyrrhotite. Dextrin, an organic colloid used to depress slimes, is thought to be a general depressant for minerals through smearing particle surfaces. Two frothers are used, cresylic acid to promote a tough high recovery froth and Aero-froth 71 which gives a cleaner froth. Cresylic acid is an industrial waste product consisting primarily of the higher homologs of the phenols. The AF 71 is produced commercially

by American Cyanamid company and consists of six to nine carbon chain alcohols. The pH of the flotation circuit is not controlled although sulphide recovery would be improved through acid addition. The ore is buffered and the pH is consistently between 7,4 and 8,5.

The modelling investigation is concerned only with the very fundamentals of flotation. Reagent effects introduce a tremendously complicated system of variables and for this reason reagent consumption and additional points were not changed during the test work.



1. Flotation of various ores



1 Chalcopyrite

2 Pentlandite

3 Pyrrhotite

4 Pyrite

5 Pyroxenite

6 Feldspar

2. Typical sulfide conglomerate<sup>6</sup>



#### IV. PRESENT FLOTATION MODELLING STATUS

The flotation process has long been regarded as more art than science. In the past, only through an extended association with the practice of flotation has the mineral processing metallurgist come to grasp its multiplicities. While in the chemical, petroleum and steel industries, mathematical modelling and sophisticated control of full-scale production processes has been well established, the mineral processing industry has remained an area where 'seat of the pants' engineering is a necessity. Flotation practice and design are still matters of experience; judgements based on the barest of facts when coupled with variations in the market of most mineral products place management personnel in a position of near impossibility in regard to production decisions. For economic optimization of a mineral processing investment, a great deal more fundamental information about beneficiation processes is necessary. Accurate information about the production capabilities of flotation works is of value, for if management can be supplied with recovery and grade estimates for varying through-puts, they will be better able to react to market fluctuations, thus maximizing profit for a fixed investment. A great deal of research is being directed towards the creation of just such a production tool; a tool designed to understand the flotation process through its fundamental laws rather than through increasing individual familiarity.

This task of increasing the degree of science in the understanding of flotation is valid, for such depth of knowledge can only lead to economic benefit. Mathematics, being the language of the sciences, is the best means to expand the logic framework of flotation. Successful mathematical modelling of flotation will aide the mineral processing metallurgist in rationalizing his flotation experience and thereby provide him with a sharp tool of competency. In addition, successful modelling

will provide production management with a sound basis for reacting to market fluctuations in an economically favourable manner.

The general objective in modelling the flotation process, i.e. development of an accurate production and design tool, may be sub-divided into several requirements for the model. The mathematical formulation must accurately describe metal and gangue recoveries in batch and continuous flotation; it must be able to predict continuous operation from the data of a batch test; it must reasonably simulate the operation of various flotation net-works, and lastly, it must respond to size and throughput variations. If a mathematical model can perform the functions with a degree of accuracy and reliability then it has become a useful production and design tool.

Attempts to mathematically define the flotation process are not entirely new; the concept of flotation as a rate process was originally proposed by Zuniga in 1935.<sup>7</sup> From the simple beginning, attempts to model flotation have burgeoned into three general categories; probabilistic, chemical analogy and linear correlation. The probabilistic approach of Schuhmann<sup>8</sup> has been developed by a great many other researchers such as Tomlinson and Fleming<sup>9</sup>, Sutherland<sup>10</sup>, Evans<sup>11</sup>, Eigeles<sup>12</sup>, Bushell<sup>13</sup>, and Arbiter and Harris<sup>14</sup>, into a basically mechanistic and complex description of collision, adhesion, retention and froth retention of particles in the flotation system. The chemical analogy models have received the attention of most researchers and also have developed through the simple single rate model<sup>15</sup>, discretely<sup>16</sup> and continuously distributed rate models<sup>17</sup>, into the complexity of the distributed constant model of King<sup>18</sup>, in which bubble residence time, particle size, available bubble surface area, aeration rate and other factors are modelled.

It is the point of view of this thesis that attempts to



develop extremely accurate and mechanistic models of flotation, as a general procedure, are unrealizable. The large number of process parameters in flotation leads the growth of such models to a degree of complexity that becomes unwieldy and impractical. Several flotation sub-processes have been described mathematically<sup>19</sup>, but an entire multitude of sub-processes defy the acquisition of accurate data as well as the subsequent mathematical formulation. If the objective of flotation modelling is the development of a production design tool, then research should be directed towards defining the limitations of the simpler models in industrial application. Only through such experimentation and feed-back can the practical goal be reached.

The purpose of this thesis then is to examine the applicability of four of the simpler chemical analogy models in the context of an industrial operation. All of the work was performed at the Frank Reduction section of Rustenburg Platinum Mines, Bleskop, Transvaal, Republic of South Africa. A model which can serve as a production and design tool for Rustenburg Platinum Mines would be beneficial, for the platinum market is historically variable. The platinum market is competitive and for this reason, disclosure of tonnages, grades, or recoveries in terms of platinum group metals (PGM) is restricted. The thesis deals only with the flotation of chalcopyrite, pentlandite and pyrrhotite; no reference is made to PGM flotation. With regard to the sulphide minerals and non-sulphide gangue, each model will be investigated as a production design tool. Any errors in the model's formulation will be examined and for the most successful model some improvements will be suggested. The degree of accuracy and reliability of each model will be illustrated and the potential worth of flotation modelling in general will be commented upon.

## V. FUNDAMENTALS OF FLOTATION AND CHARACTERISTICS OF THE MODELS

In order to evaluate a mathematical model of flotation it is necessary to understand both the fundamental characteristics of flotation and the basic structure of the model. In this chapter, the fundamentals of flotation, the basic structure of the chemical analogy models and the expected limitations of the four models will be discussed. Many of the possible pitfalls of model testing will be illuminated and any inconsistencies between the model structure and flotation fundamentals will be examined.

The number of variables in flotation is extremely large as shown in Table I.<sup>20</sup> Simple mathematical models such as are to be examined here can only hope to account for several of the major flotation process characteristics. An effort to incorporate all flotation parameters involves a complexity, probably without increased accuracy, beyond imagination. Hence, the thesis is concerned with an examination of only the fundamental flotation variables and the ability of the model to simulate these fundamentals.



TABLE I.  
FLOTATION VARIABLES

Uncontrolled	Controllable
Raw Material	
Nature of valuable mineral Secondary valuable minerals Associated gangue Soluble constituents of the ore Degree of oxidation of the ore Oxidation during mining Quantity of make-up water available	Soluble impurities in the water Presence of dissolved gases Degree of alkalinity or acidity of water used
Grinding and classification	
Fineness of particles and closeness of association Degree of hardness of minerals and gangue Relative time for different minerals in grinding circuit	Type of machine used Type of grinding media used Oxidation during grinding Chemicals added during grinding Pulp density during grinding Rapidity of circulation Time
Conditioning	
	Pulp density Chemicals added Order and form of adding the chemicals Temperature Mean time allowed Intensity of agitation
Flotation	
Particle density	Particle size Pulp density Temperature pH value Circulating load Time in the machine Geometry of the machine Power intensity Impeller speed Degree and type of aeration Height of froth Specific chemicals added to circuit

### The Flotation Process

The flotation process is an industrial method for separating small mineral particles from one another. The separation is performed in a water suspension of the mineral particles concerned; air is dispersed throughout this suspension and by suitable chemical treatment of the slurry, certain mineral surfaces are rendered hydrophobic. The hydrophobic mineral particles adhere to the rising air bubbles and are removed in a soapy froth. Obviously the object is to choose the chemical conditioning stage such that the valuable mineral particles are removed from the non-valued minerals.

In industrial application, flotation occurs in a series of two port tanks, the concentrate froth is removed at a variable rate from the tank and the tailings flow continuously to the next unit. Often a number of recycle streams are employed in an effort to improve the grade of the final product. Through a series of flotation cells the tailings stream is successively depleted of its more floatable, air-avid material and the concentrates are enriched in floatable, usually valued minerals.

From the micro rather than the macro view point, the mechanism of flotation becomes considerably more complex. The flotation of each individual valuable mineral particle, involves the complex chemistry of activation and collector coating as well as the physical processes of particle-bubble collision, bubble-particle adhesion and particle removal in a stable froth. It was essentially these physical micro processes which concerned early researchers like Schuhmann<sup>21</sup> who established the definite probabilistic nature of flotation. One of the major characteristics is the bubble-particle collision and adhesion, often this is considered similar to a reaction between particles and bubbles. This probabilistic



or time dependent nature of flotation is a primary characteristic of any flotation process.

A second physical sub-process, the removal of particles in a stable froth, has been mentioned as a significant factor by Arbiter and Harris,<sup>22</sup> and will be later shown to have a tremendous effect upon recovery and grade in a flotation cell. This second fundamental characteristic of the flotation process may be viewed as interacting with the time dependency of flotation; as the froth removal rate increases so does the apparent rate of flotation.

The third and fourth primary characteristics of the flotation process are more concerned with the particles themselves rather than any physical sub-processes in flotation. Firstly, it must be realized that each particle within a flotation pulp is a complete individual. Even those particles which may be classed as part of a mineralogical or metal species vary greatly in size, shape and surface characteristics, it would be miraculous indeed if particles of such variety exhibited identical flotation properties. The fourth factor which is of tremendous importance in application of the chemical analogy flotation models is brought about by the use of recycle streams within the normal industrial flotation network. Due to the inclusion of several recycle streams many mineral particles may circulate within the circuit for a considerable time before being removed at either the final concentrate or final tailings streams. Because flotation is basically a surface phenomena, the floatability of such particles may be effected by oxidation, collector coating removal or attrition while within the circuit. The significance of this factor is yet to be determined but such a time dependency of the particles' flotation characteristics must be noted as a possibility.



### Structure of the Chemical Analogy Models

Discussion of modelling structure will be limited to the structure of the four models investigated at Rustenburg Platinum Mines. Throughout this thesis the four models will be referred to as 1) the single rate model, 2) the rate plus non-floating model, 3) the gamma function model, and 4) the two-phase model. Each of these four models is based upon first-order kinetics; that is the process of flotation is assumed to be consistent with a first-order reaction process. In the case of the two-phase model, both the froth and pulp phases are assumed to be governed by first-order kinetics. The general equation for a first-order reaction is:

$$\frac{dC}{dt} = -kC$$

where 'C' is the reactant concentration and 'k' is the first-order rate constant.

All of the models investigated assume this simple description of flotation as a first-order process and assume that rate constant values for a particular group of particles remain constant throughout a flotation network. The models differ in their distribution of first-order rate constants for a mineralogical or metal species and in the number of phases which are mathematically incorporated in the model structure. The single phase models, that is, the single rate model, the rate plus non-floating model and the gamma function model assume that the froth phase may either be neglected or may be incorporated as a rate determining factor, rather than as a separate phase. The single rate model assumes that the flotation of a mineralogical or metal species may be described by a single rate constant value. The rate plus non-floating model recognises the existence of a non-floatable portion of material

within every mineralogical or metal species, thus it describes flotation in terms of two first-order rate constants, one of which is zero. The gamma function model recognises the inherently individual nature of the multitude of particles in a particular mineralogical or metal species and assumes a modified form of the gamma function as a continuous distribution of first-order rate constants. The continuous rate constant distribution is entirely defined by its mean and variance. The two-phase model investigated, incorporates the froth phase directly in its basic structure and acknowledges the obvious return of solids to the pulp from the froth phase.<sup>23</sup> The transfer of a mineralogical or metal species between the two phases is governed by two rate constants; one first-order constant for transfer from pulp to froth and one first-order constant for transfer from froth to pulp.

As with many processes, flotation may be performed either as a batch or as a continuous process. In any time dependent process these two cases of operation are analytically dissimilar. In the batch model of flotation each particle is susceptible to flotation conditions for an equal length of time; in continuous operation however, a distribution of solids residence times exists. A theoretical residence time distribution equivalent to 'perfect mixing' is often assumed. It will be later demonstrated that the assumption of such a distribution is reasonable at Rustenburg.

#### Possible Limitations in the Model Structure

In order to understand the flotation process and expand that understanding through mathematical modelling of the process, it is necessary to clearly expose the assumptions of any model and discuss the limitations and possible problems which might arise in the application of that model. In the



two previous sections, the primary characteristics of industrial flotation and the basic structure of the investigated flotation models were discussed. It now remains to examine the plausibility of the basic model structures in regard to the flotation process primary characteristics.

All first-order models have assumed that the rate constant for a particular particle class is time-invariant; both with respect to varying mineralogies in the feed to the flotation network, and throughout the recycling in an industrial circuit. The various mineralogies at Rustenburg Platinum Mines do not float with the same rate distribution as previously illustrated in figure 1. Varying proportions of these mineral bearing bands in the flotation feed will alter the rate constant distribution for any particular mineral or metal species. It has not been demonstrated that the individual particles within a flotation network maintain their floatability throughout the various re-cyclings and re-flotations in the normal industrial circuit. Even if the normal practice of staged reagent additions is ignored, the inevitable processes of attrition and surface oxidation must affect the floatability of particles in the system. In the re-flotation of concentrates, the particles' floatability may well be enhanced by the previous attachment of bubbles to their surfaces. The extent of this modelling assumption is not yet determined. Obviously the seriousness of feed variation depends both on the frequency and amplitude of such changes; the maintenance of a particles' floatability may or may not be of significance with regard to the objectives of flotation modelling but it is necessary that it is realized that the first-order models do not cater for and could not cater for a variation in the rate constant with time without the addition of many complicating mathematical features.



The froth phase and froth removal rate will be later demonstrated to be a very significant factor in flotation. Arbiter and Harris<sup>24</sup> have correctly described the fundamentals in that the flotation process is essentially a two-phase system, a pulp phase and a froth phase. Without some means other than direct kinetic consideration of the froth phase, the three single phase models can be expected to be severely limited by the assumption of a single phase system.

In consideration of the very individual nature of particles in a flotation pulp, it must be realized that the discrete rate constant models can only approximate the real situation. The gamma function model possesses a distinct advantage over the other three models investigated as it assigns a continuous distribution of rates to any species. In a recycle system, small errors arising from the assumption of discrete rate constants may accumulate into large errors in the entire system.

Any discrete rate constant or distribution of rate constants is obviously reactant specific. Throughout the application of a model it is imperative that material representative of the feed to the flotation network is examined. It became clear early in the research work that standard laboratory batch testing of plant feed material was considerably different to batch testing of plant pulp samples. It cannot be expected that material which is crushed and ground in a laboratory batch ball mill will give rise to particles with flotation characteristics which are identical to the characteristics of the same material comminuted in several stages of plant milling and cyclone separation. Figures 3, 4, 5 and 6 are comparisons of laboratory batch tests and batch tests done on the plant feed stream which underline the possibility of gross discrepancies between essentially the same material prepared for flotation in the laboratory and in the plant. The major discrepancy

lies with the pyrrhotite mineral fraction. It is plausible that due to the ease of oxidation of the iron sulphide surfaces,<sup>25</sup> the higher plant milling temperatures have to a greater extent oxidized the pyrrhotite surfaces and reduced their floatability. Evidence of considerable iron sulphide oxidation is shown in figure 7, where acid addition has temporarily lowered the pulp pH and removed the oxide film on the sulphide surfaces, even after the pulp pH is returned to normal the iron sulphide flotation is enhanced. A second factor in the realization that flotation rate constants are reactant-specific became evident during research and is concerned with the quality of water used for flotation. Various ion species are known to adversely affect the flotation of sulphides<sup>26</sup> and no less can be expected as flotation is a surface phenomena. In illustration, figure 8 shows the effect of a change in plant water quality upon the final tailings value of the investigated concentrator. Due to the publication restrictions of the Rustenburg Platinum Mine, no actual value may be given to the final tailings grade, however, a considerable effect is evident.

These two previous illustrations serve only to illuminate the complexity of the flotation process and outline the necessity for careful consideration of metallurgical factors in application of any flotation model to an industrial plant.

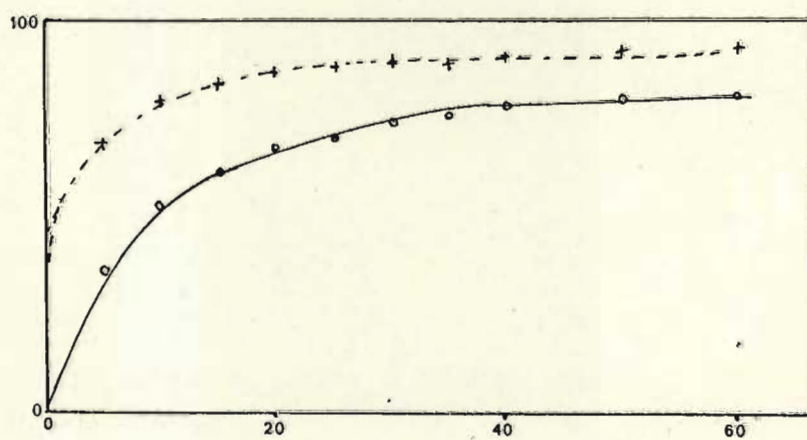
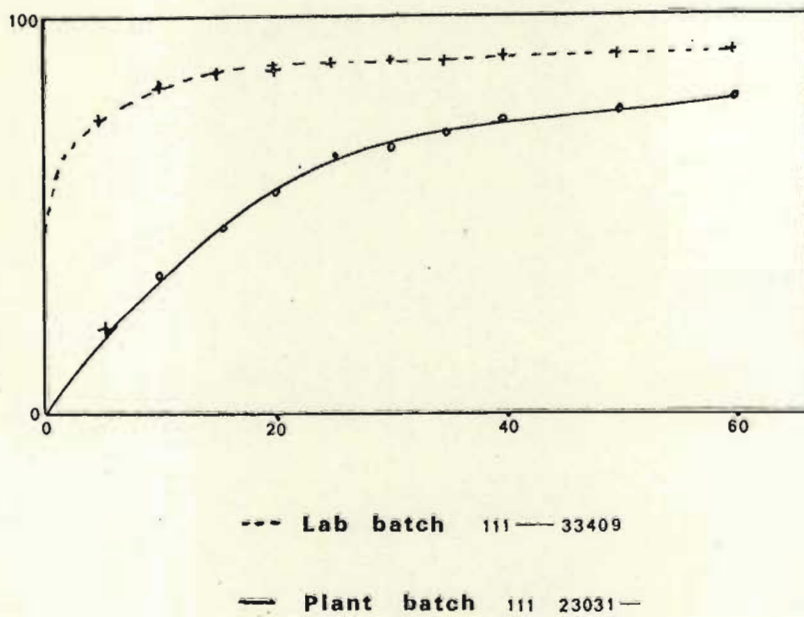
It must also be born in mind with regard to first-order models as investigated at Rustenburg, that the rate constant or rate constant distribution for a mineral or metal species is equipment-specific. Such variables as impeller speed and design, air suction, frothing, surface to volume ratios, the means of froth removal and agitation intensity may have considerable effect upon the flotation rates of various minerals. As large scale plant cells are difficult to scale down to a reasonable size for batch operations, some discrepancies will obviously



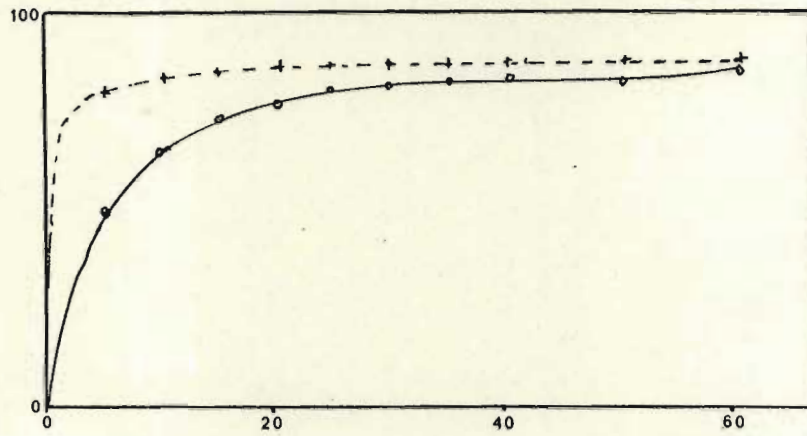
exist between flotation equipment in batch and continuous testing work. The assumption that any batch test may be related to continuous operation through a residence time distribution may be seriously in error, and hence will contribute to the predictive quality of kinetic parameters regressed from batch testing.

The gamma function model is inherently the most powerful of the four investigated models for it incorporates the time-dependent nature of flotation; it recognizes the individual nature of the mineral particles and can easily simulate the complex industrial networks of most mineral processing plants. Provided that the froth phase can be included in any simulation as a factor which affects the rate constant distribution, it will provide the best basis for building a production design tool. Although a number of assumptions have been made in the application of all the models including the gamma function model, it must be remembered that the objective of the flotation modelling exercise is not the development of an every way complete description of flotation but rather the development of a simple model which can be used with reasonable accuracy and confidence as a tool for production and circuit design analysis of existing flotation networks.



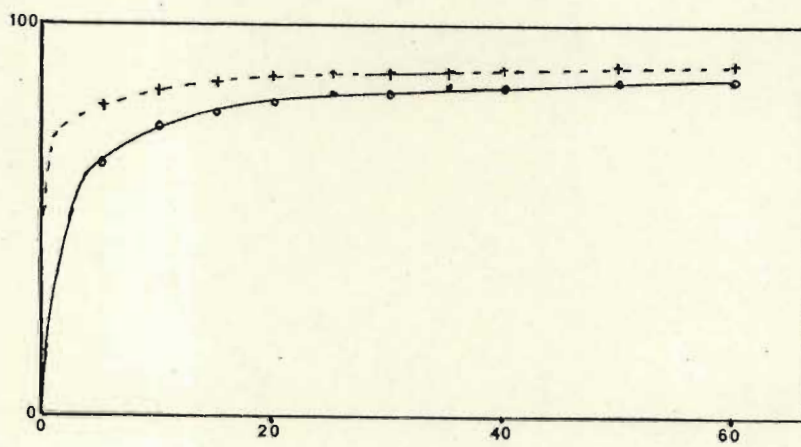


3. Sulfur recovery (%) versus  
flotation time (mins)



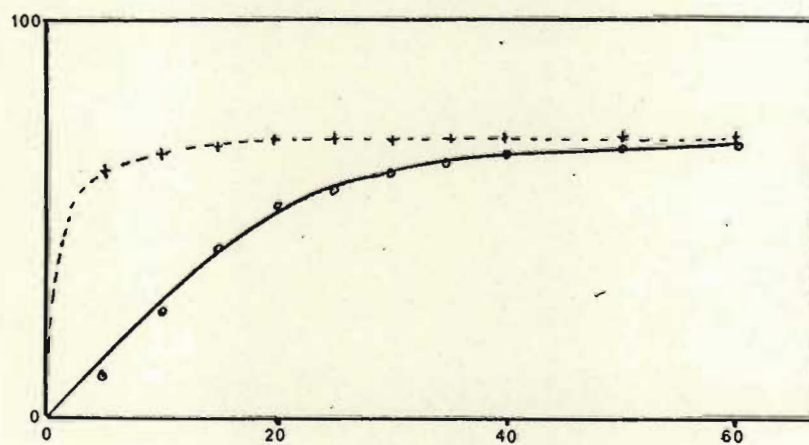
--- Lab batch 131 — 33409

— Plant batch 131 23031 —



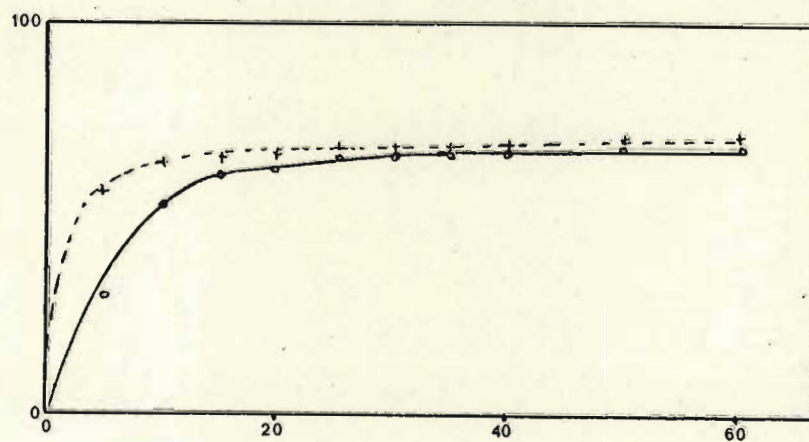
4. Chalcopyrite recovery (%)

versus flotation time (mins)



--- Lab batch 141-33409

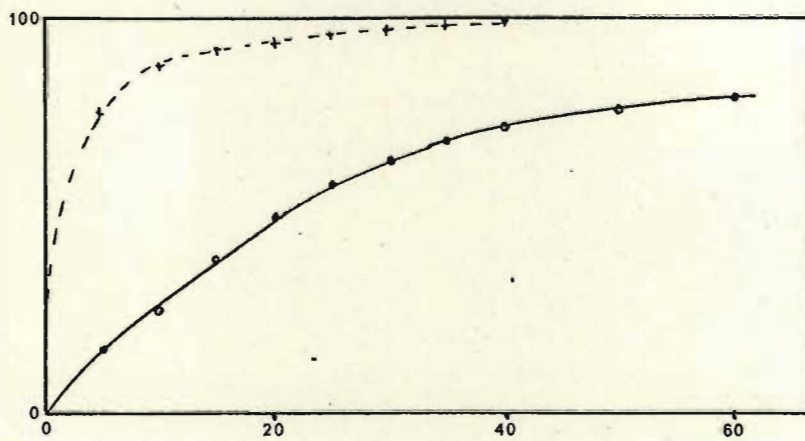
— Plant batch 141-23031 —



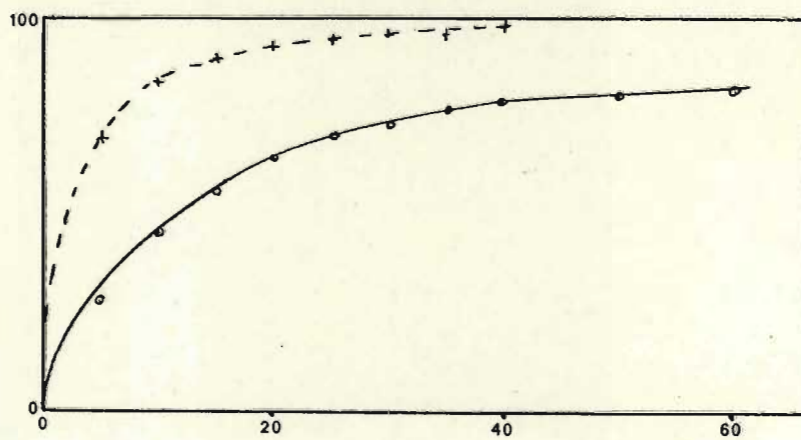
5. Pentlandite recovery (%)

versus flotation time (mins)



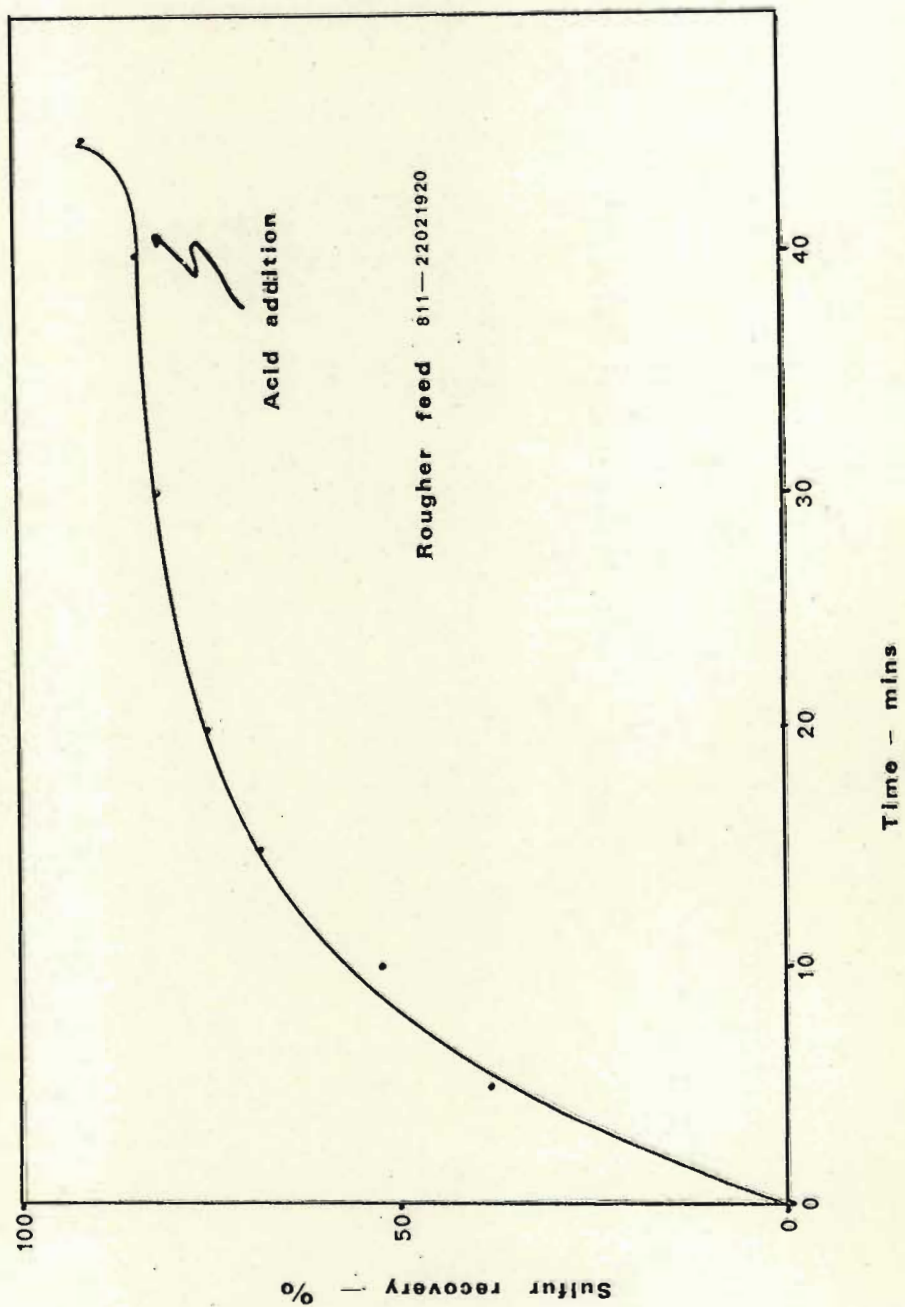


--- Lab batch 151 33409  
— Plant batch 151 23031 —

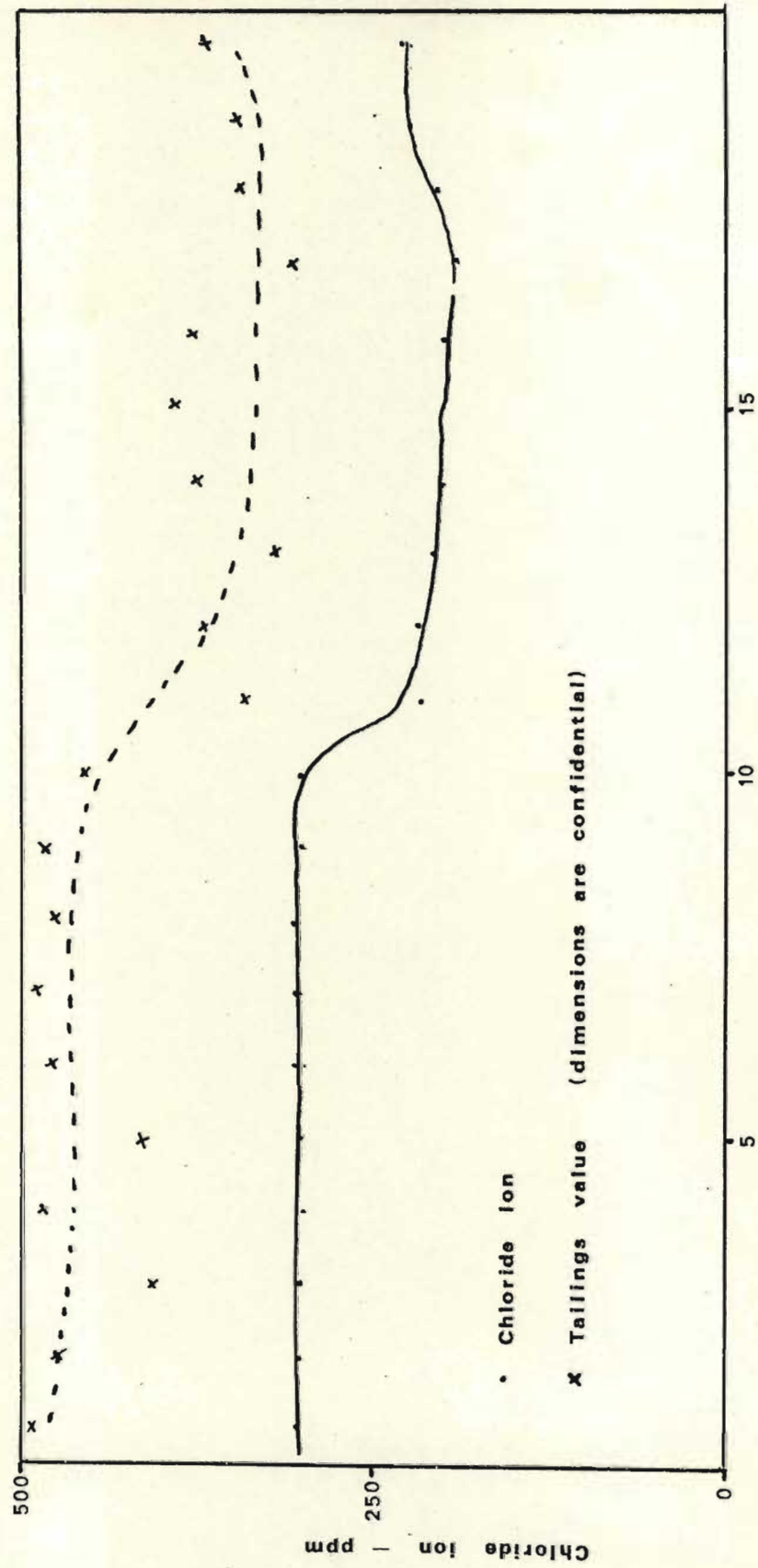


6. Pyrrhotite recovery (%)

versus flotation time (mins)



7. Acid addition improves flotation



Number of observations - not necessarily at equal intervals

8. Effect of changes in mill water



## VI. APPLICATION OF THE FOUR FLOTATION MODELS

In the analysis of the predictive quality of a number of flotation models with regard to industrial operation, several sub-goals may be readily defined. As an initial requirement of a valid flotation model it must be accurately descriptive of batch cell recovery vs. time curves. This is a relatively simple requirement and the majority of flotation models will fit batch data with reasonable accuracy.

The second more stringent test of a model's predictive capacity is also readily available in an operating plant. If the feed stream to an industrial flotation bank is sampled and floated in a batch cell, the kinetic parameters regressed from the batch curve may be used to predict the performance of the continuous bank. A recovery curve for the continuous operation may be obtained through sampling and compared to the model prediction. It is primarily in this testing that any problems arising from the fact that the kinetic parameters are equipment-specific will be evident.

One of the modelling assumptions which may be criticized is the time-invariant nature of the kinetic parameters. Considerable scope is available for testing this assumption in an operating plant, at Rustenburg this assumption was tested by performing batch tests on the rougher concentrates in connection with batch tests of the rougher feed. In theory, the batch flotation of the rougher feed will define the rougher plant operation and the consequent batch flotation of the rougher concentrate.

The fourth and most crucial test of the model's ability to serve as a production and design tool is based on the model's ability to predict the flotation operation in various circuit arrangements. A number of flotation networks were operated and

sampled for later comparison with the model predictions.

These four tests of the model applicability should well define any shortcomings in the model structure as well as illustrate the degree of confidence which may be placed in model simulations.

Throughout the thesis discussion, all raw data is referenced by a fourteen-digit code number; the raw data and full details of the code meaning can be found in the first section of part two of the thesis.

#### Measurement of the Residence Time Distribution

As previously discussed, in a continuously operating flotation cell each solid particle is subjected to a different flotation time depending upon its exact course of travel through the cell. This distribution of residence times which exists for the solids feed to the cell must be experimentally determined and mathematically formulated before the kinetic parameters of batch flotation may be applied to continuous operation. Jowett<sup>27</sup> has tested the assumption that a flotation cell operates as a perfect mixer and reported reasonable accuracy. Colborn<sup>28</sup> has reported a significant variation in residence time distributions for various size fractions. At Rustenburg, a liquid tracer test was performed on a bank of rougher cells and on a bank of cleaner cells. Sodium chloride was added to the head of the flotation bank in crystalline form; the tailings solution was then analyzed for chloride ion by titration with silver nitrate; potassium dichromate served as an indicator. Loveday<sup>29</sup> has presented the equation for the residence time distribution of a series of 'n' perfectly mixed cells



as follows :

$$E(t) = \frac{(nv/V)^n}{(n-1)!} t^{n-1} e^{-ntv/V}$$

where 'V' is the effective volume of a single stirred unit and 'v' is the volumetric flowrate to the cells. Thus from the residence time distributions obtained, the effective volume of a single unit may be estimated. The residence time distribution obtained and the regressed fits of the perfectly mixed equation are shown in figures 9 and 10. Considerable discrepancy may exist between solid and liquid residence time distributions, particularly solids of various sizes. In addition to the liquid residence time tests performed at Rustenburg, solids radioactive tracer tests on four size fractions in the rougher bank were performed by personnel of the Atomic Energy Board.<sup>30</sup> Their results are summarized in Table II below :

<u>Tracer</u>	<u>Regressed Effective Volume of Unit</u>	<u>Date</u>
Cl <sup>-</sup>	622 litres	7-1-74
+ 208 μ * (radioactive)	537	19-3-74
+ 104 μ * "	551	19-3-74
+ 53 μ * "	563	19-3-74
- 53 μ * "	571	19-3-74

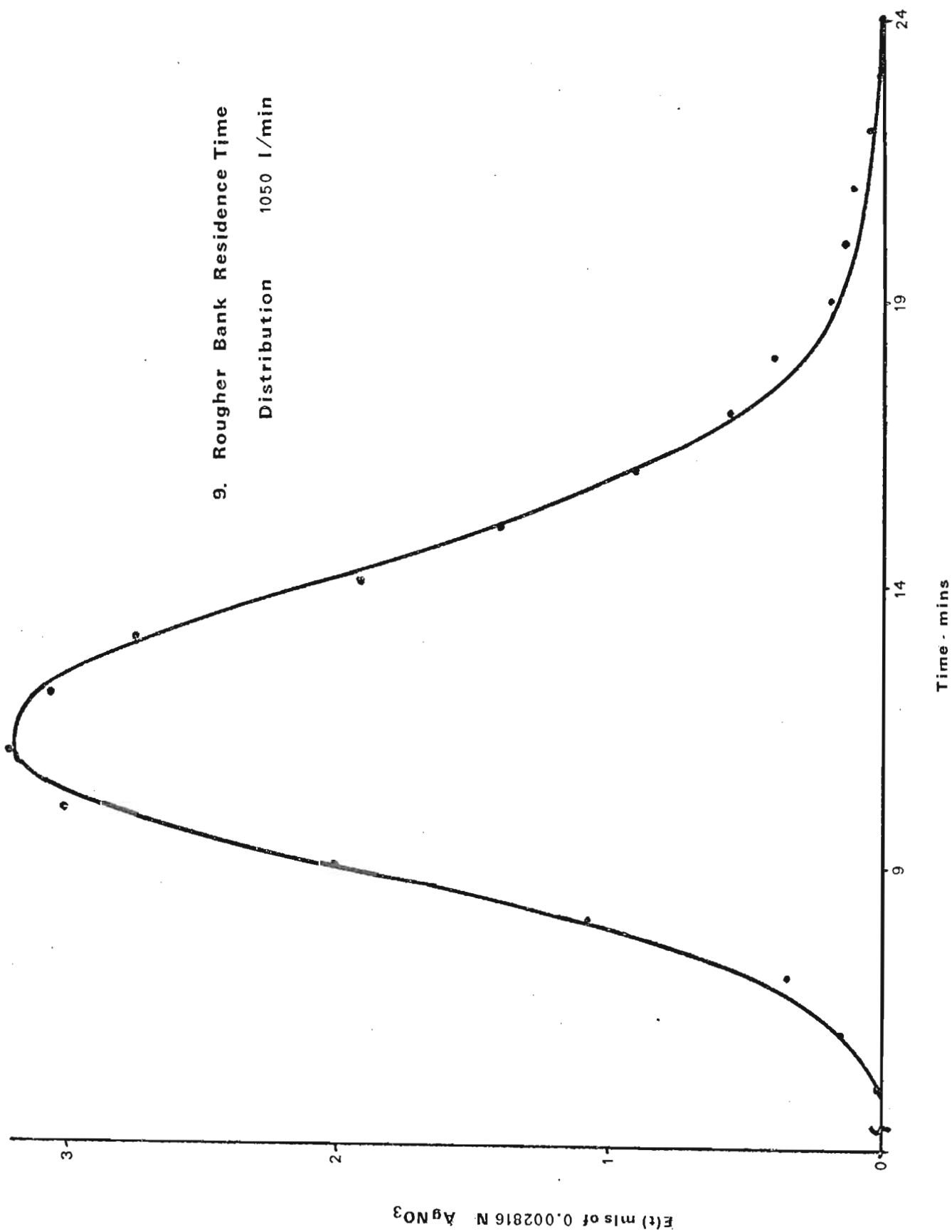
\* performed by Atomic Energy Board personnel

Although the solids effective cell volumes are slightly lower than the liquid testing shown, there is little variation between size fractions. It is assumed that operating conditions of froth depth and froth removal rates can give rise to significant variations in the effective cell volumes hence the volumes of 622 l/cell were used throughout the thesis as the

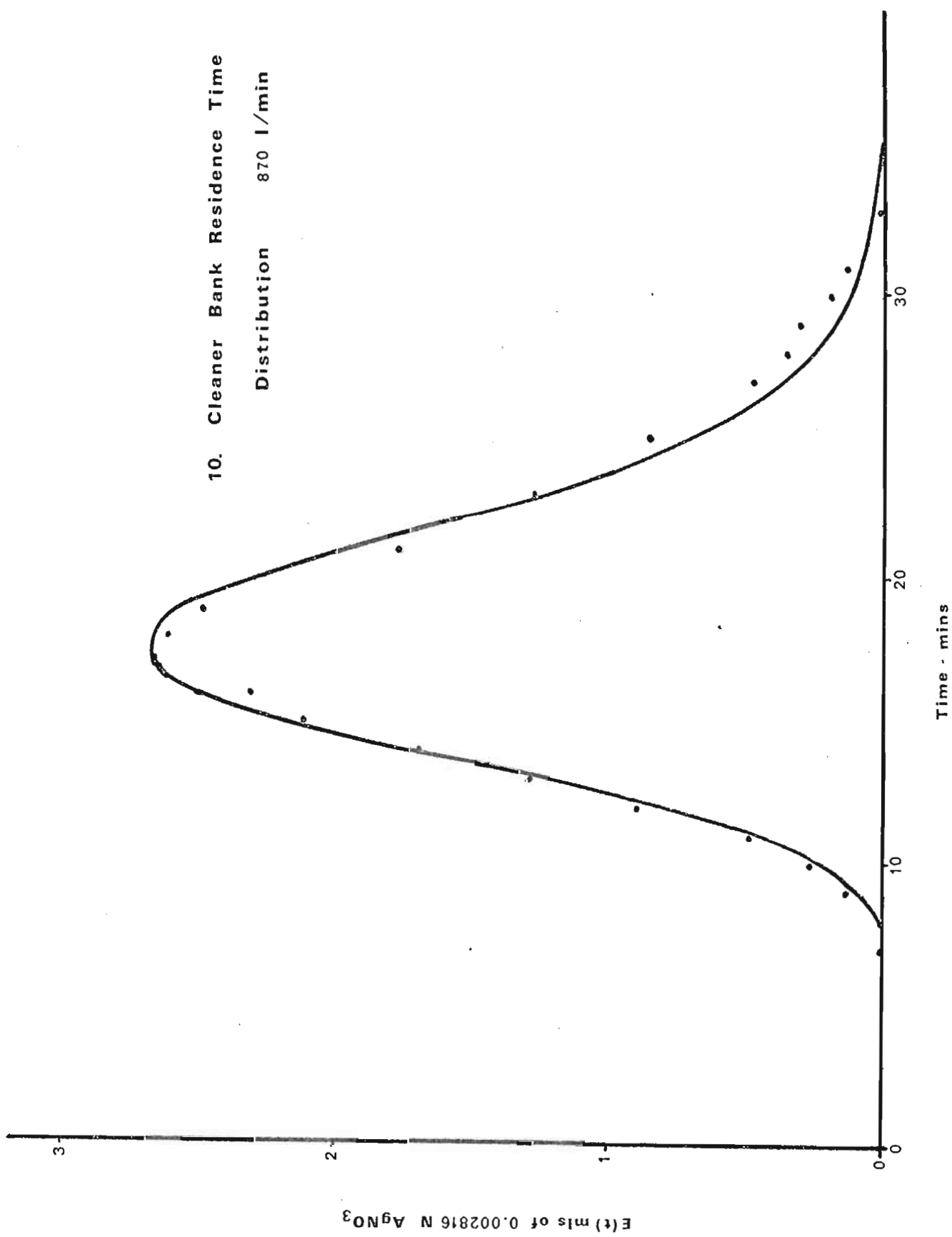
effective perfectly mixed volume of any flotation cell. It is worthwhile to note that a considerable discrepancy exists between the effective cell volume and the cell volume calculated from its dimensions; the effective volume is approximately fifty-five percent of the manufacturer's dimensional volume.



9. Rougher Bank Residence Time  
Distribution 1050 l/min



10. Cleaner Bank Residence Time  
Distribution 870 l/min



## PERFORMANCE OF THE SINGLE PHASE MODELS

Each of the single phase models may be used in theory to describe batch flotation, continuous flotation, or flotation in a plant network. Because of their varying treatment of the distribution of first-order rate constants, the single rate model, the rate plus non-floating and the gamma function model describe each flotation circumstance with varying degrees of accuracy. Initially the first three sub-goals outlined for the analysis of the predictive quality of the single-phase models will be discussed separately for each model. The models may be expected to have reasonable descriptive capacity for batch flotation and a good measure of predictive capacity in a simple single bank continuous operation. The prediction of concentrate refloatation can be expected to illuminate problems in the model structures. The single phase models can also be expected to be severely limited in application to various frothing conditions.

### The Single Rate Model

The single rate first-order irreversible model proposes that the flotation of a species is a single exponential function of time. Every particle in a designated species floats according to the same first-order rate constant.

A large number of batch tests were performed on various plant streams at Rustenburg; species were described through physical and chemical means. Several examples of the first-order model's ability to fit observed batch data for a species are given in Figures 11, 12 and 13. It is obvious from the figures that the single rate model is considerably lacking in its descriptive powers.

Theoretically if the flotation of a species may be



described by a single first-order rate constant, then the batch flotation of any metallurgical stream in a flotation network would give rise to the same regressed rate constant for a particular species. This is definitely not true as can be seen in Table III. The same flotation species in various streams floated in the same batch equipment gives rise to considerably different first-order constants.

The single rate model may also be applied to flotation in a series of continuous flotation cells; fits of the model to continuous operation concentration-time curves are shown in Figures 14 and 15. Again the same problems of inaccuracy are evident in the continuous flotation case.

It is obvious from a preliminary study of batch concentration curves that invariably some portion of any mineralogical or metal species remains non-floatable at long flotation times. Thus the single rate model could be improved by the mathematical recognition of this unfloatable fraction.

TABLE III. SIMPLE FIRST ORDER RATE CONSTANTS

<u>MATERIAL AND TEST CODE</u>	<u>FLOTATION STREAM</u>	<u>RATE CONSTANT min<sup>-1</sup></u>
SULPHUR		
31146210911711	Rougher Concentrate	0,08198
41146210911515	Cleaner Feed	0,08666
61144070911610	Cleaner Tailings	0,02905
61146210912019	" "	0,01934
51144070911850	Cleaner Middlings	0,1024 *
51146210911625	" "	0,04751 *
81144070911700	Rougher Feed	0,02379
81146210911912	" "	0,02511
NON-SULPHIDE		
32146210911711	Rougher Concentrate	0,01734
42146210911515	Cleaner Feed	0,02057
62144070911610	Cleaner Tailings	0,01088
6246210912019	" "	0,006152
52144070911850	Cleaner Middlings	0,05173 *
52146210911625	" "	0,02145 *
82144070911700	Rougher Feed	0,001691
82146210911912	" "	0,001031
COPPER		
431--110411445	Cleaner Feed	0,2094
831--130411435	Rougher Feed	0,1642
831--230411200	" "	0,1465

TABLE III. (Continued)

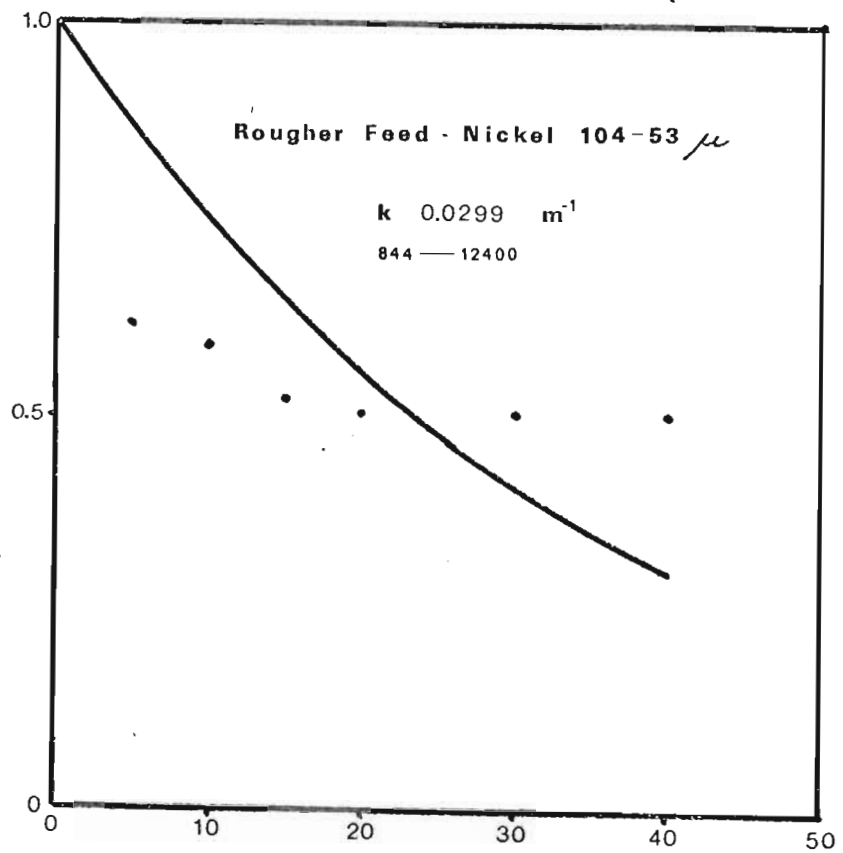
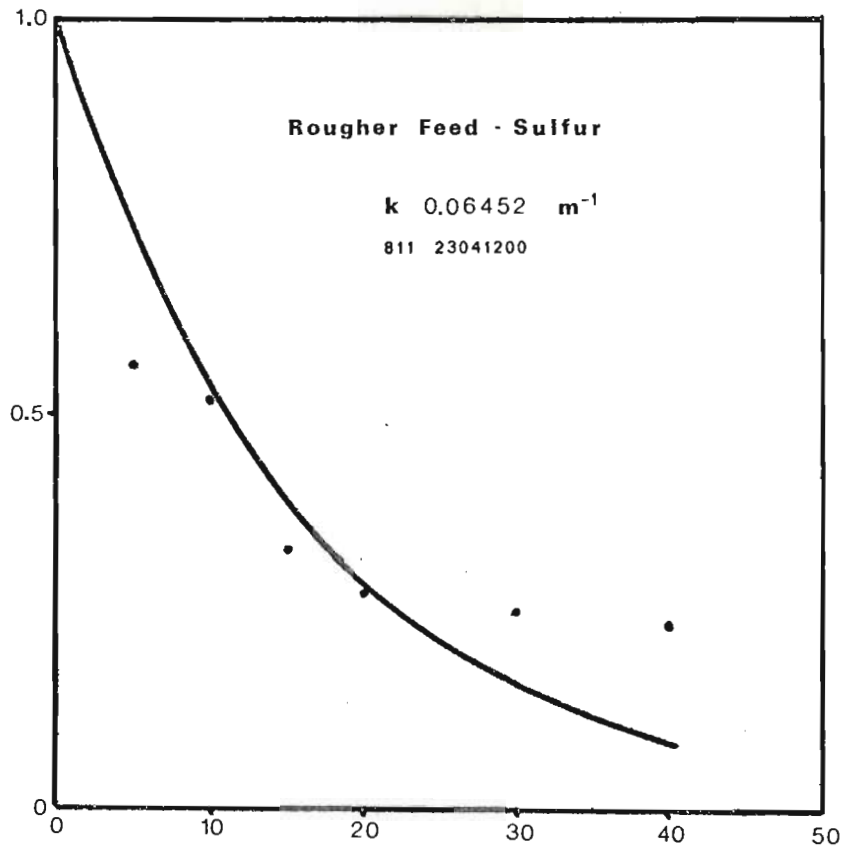
<u>MATERIAL AND TEST CODE</u>	<u>FLOTATION STREAM</u>	<u>RATE CONSTANT min<sup>-1</sup></u>
COPPER (208-104 microns)		
333-----12400	Rougher Concentrate	0,3084
833-----12400	Rougher Feed	0,03866
COPPER (104-53 microns)		
334-----12400	Rougher Concentrate	0,4723
834-----12400	Rougher Feed	0,1488
COPPER ( - 53 microns)		
335-----12400	Rougher Concentrate	0,4933
835-----12400	Rougher Feed	0,04103
NICKEL		
441--110411445	Cleaner Feed	0,1108
841--130411435	Rougher Feed	0,05895
831--230411200	" "	0,06819
NICKEL (208-104 microns)		
343-----12400	Rougher Concentrate	0,1171
843-----12400	Rougher Feed	0,007798
NICKEL (104-53 microns)		
344-----12400	Rougher Concentrate	0,2517
844-----12400	Rougher Feed	0,02990



TABLE III. (Continued)

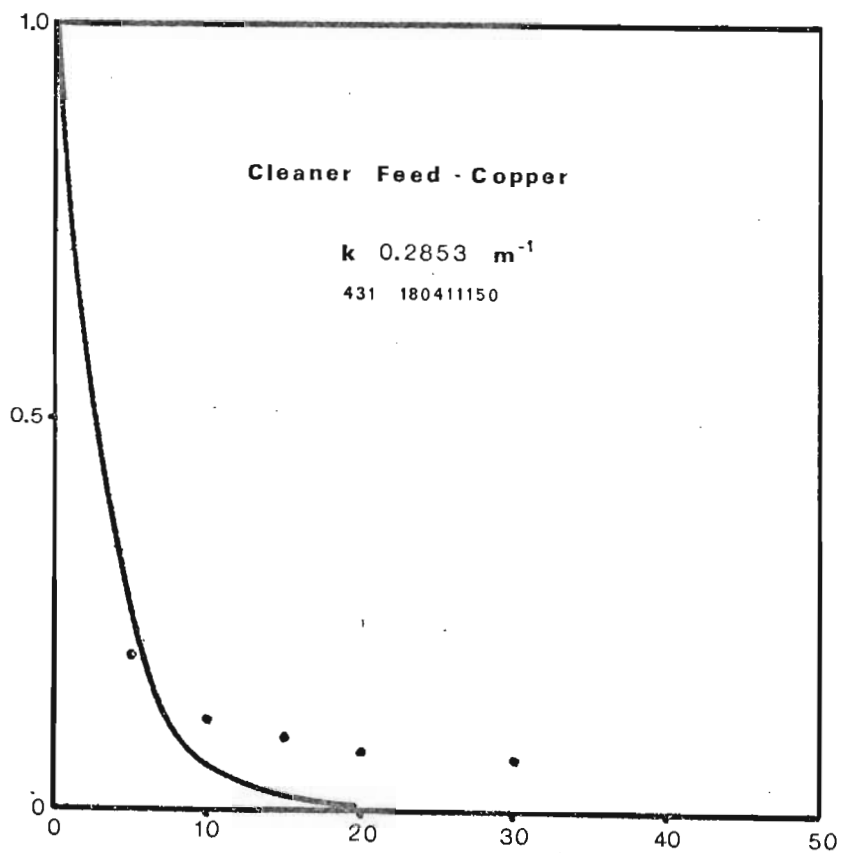
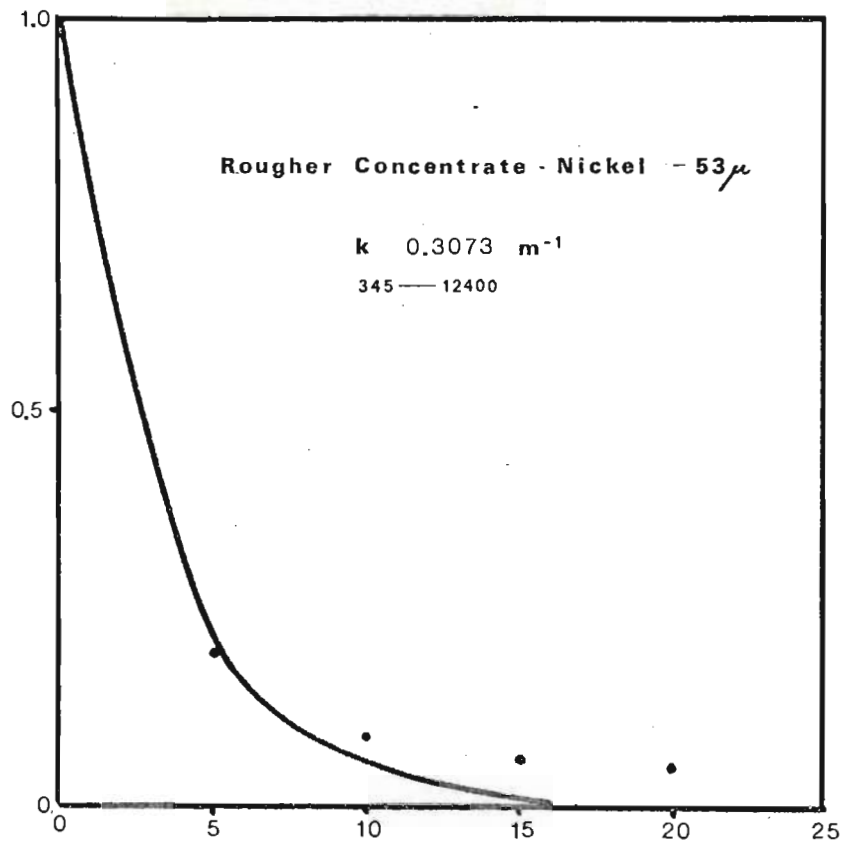
<u>MATERIAL AND TEST CODE</u>	<u>FLOTATION STREAM</u>	<u>RATE CONSTANT min<sup>-1</sup></u>
NICKEL ( - 53 microns)		
345-----12400	Rougher Concentrate	0,3073
845-----12400	Rougher Feed	0,03843

\* These two tests were performed at greatly different froth removal rates.



# 11. Batch Tests - Single Rate Model

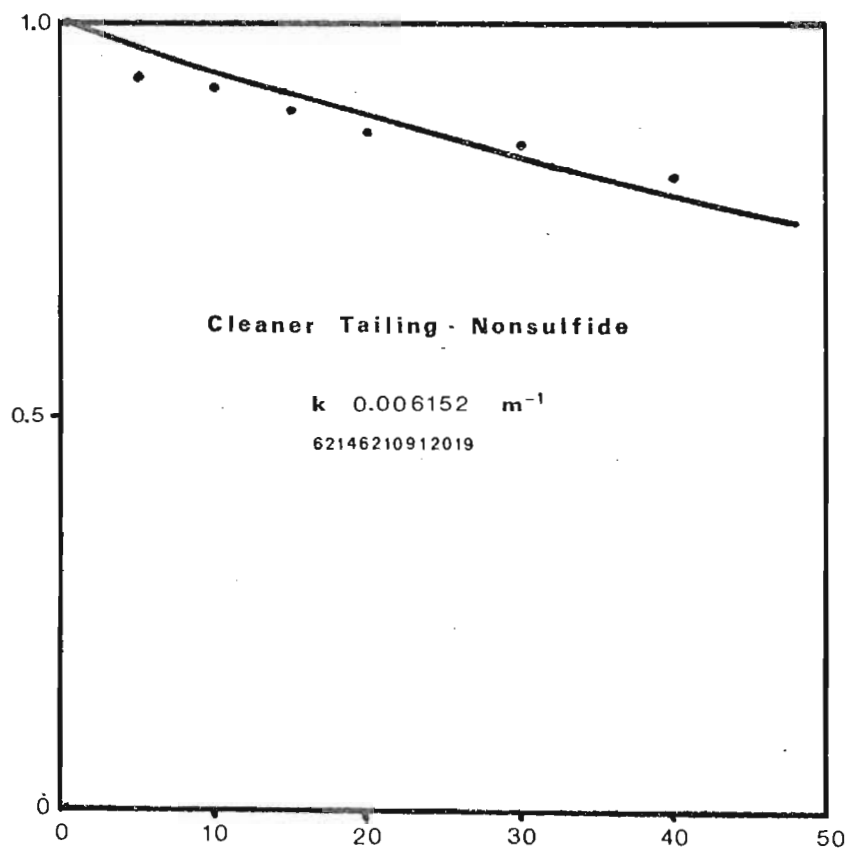
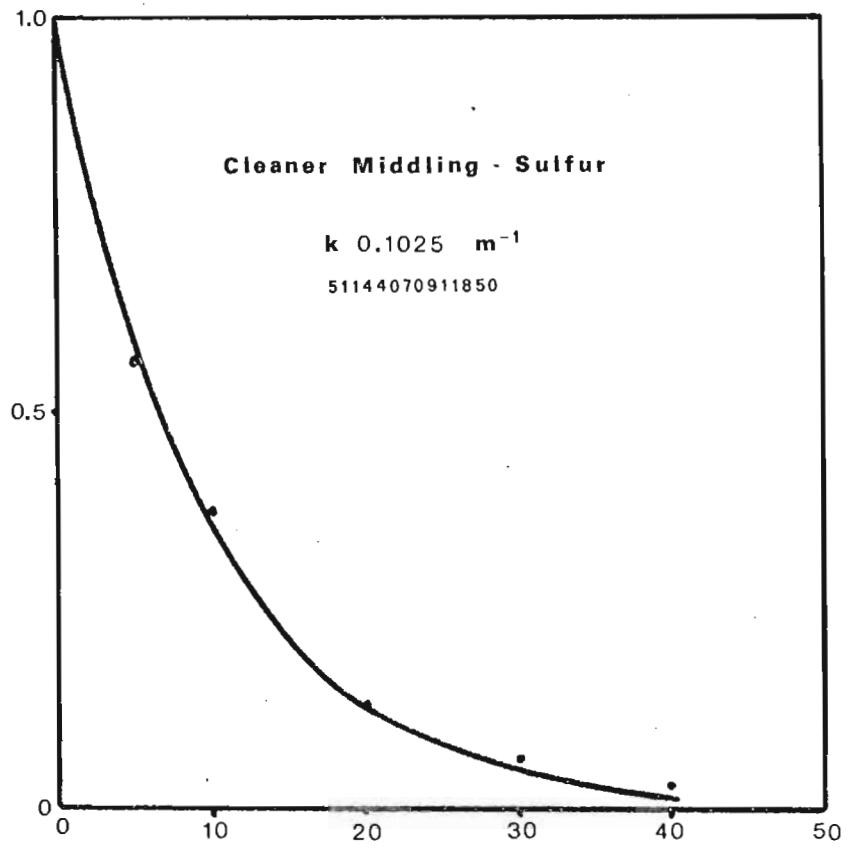
Fraction in Tailings vs Time - mins



**12. Batch Tests - Single Rate Model**

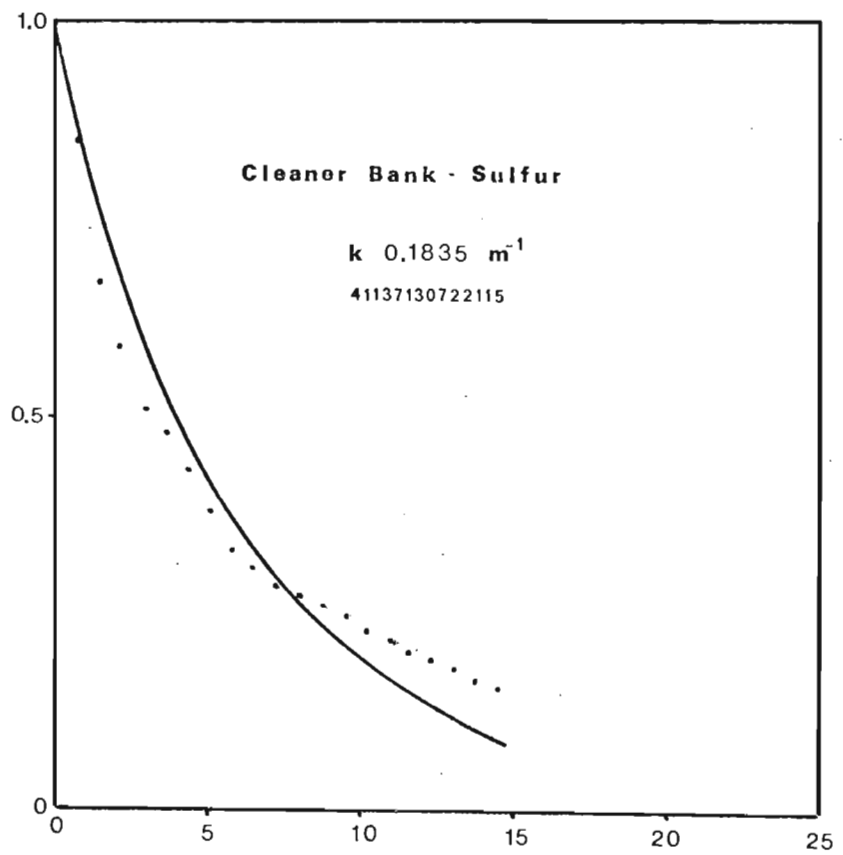
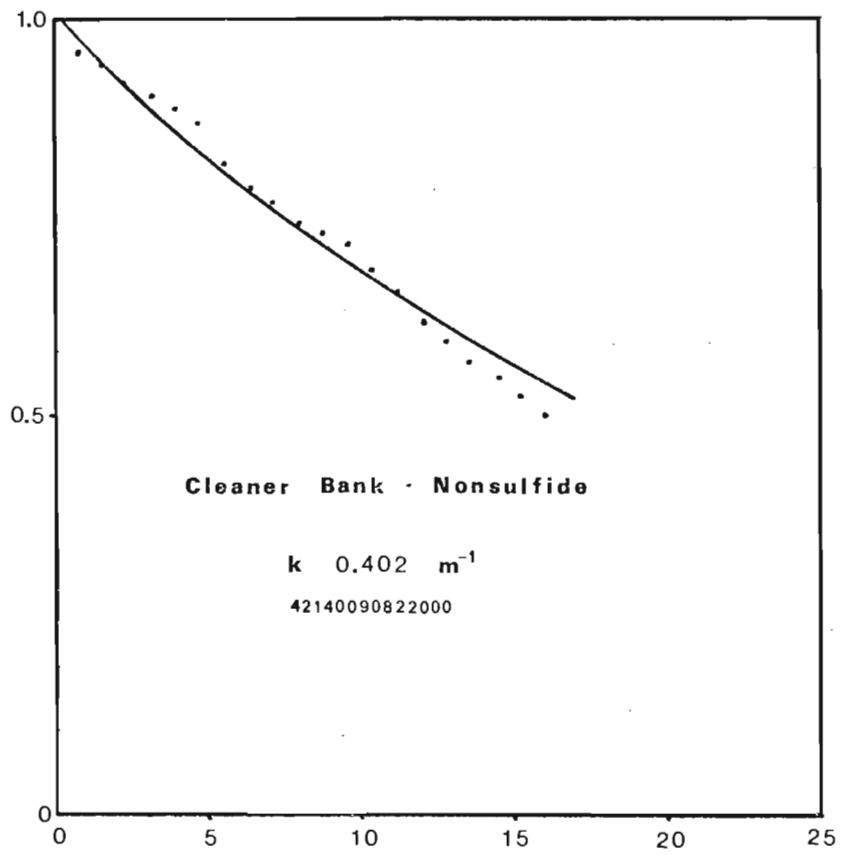
**Fraction in Tailings vs Time - mins**





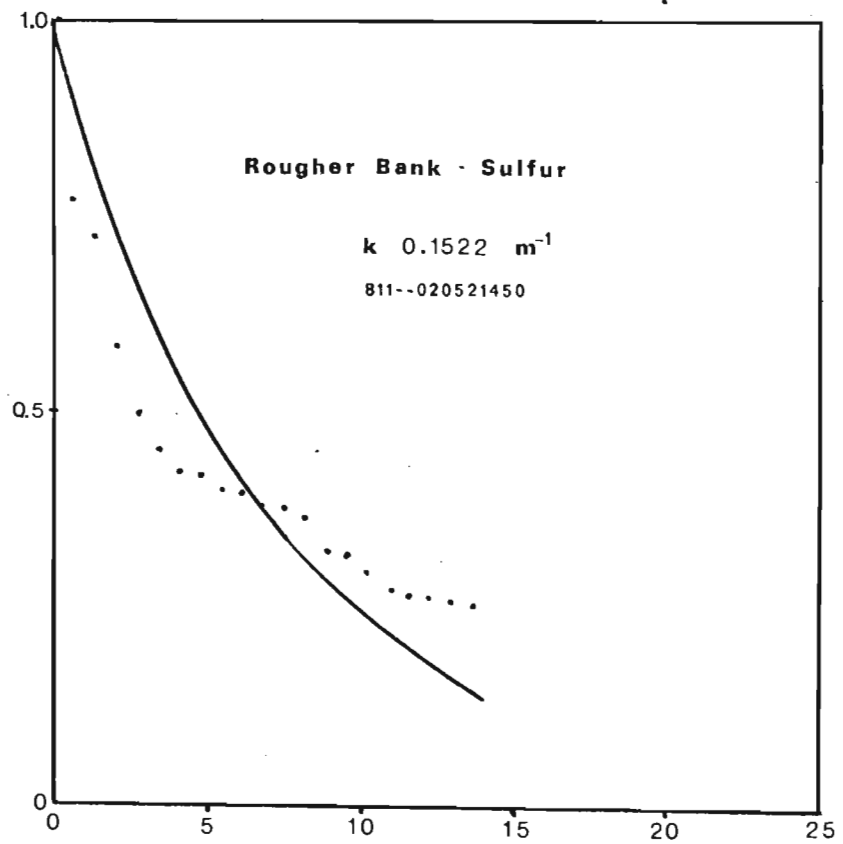
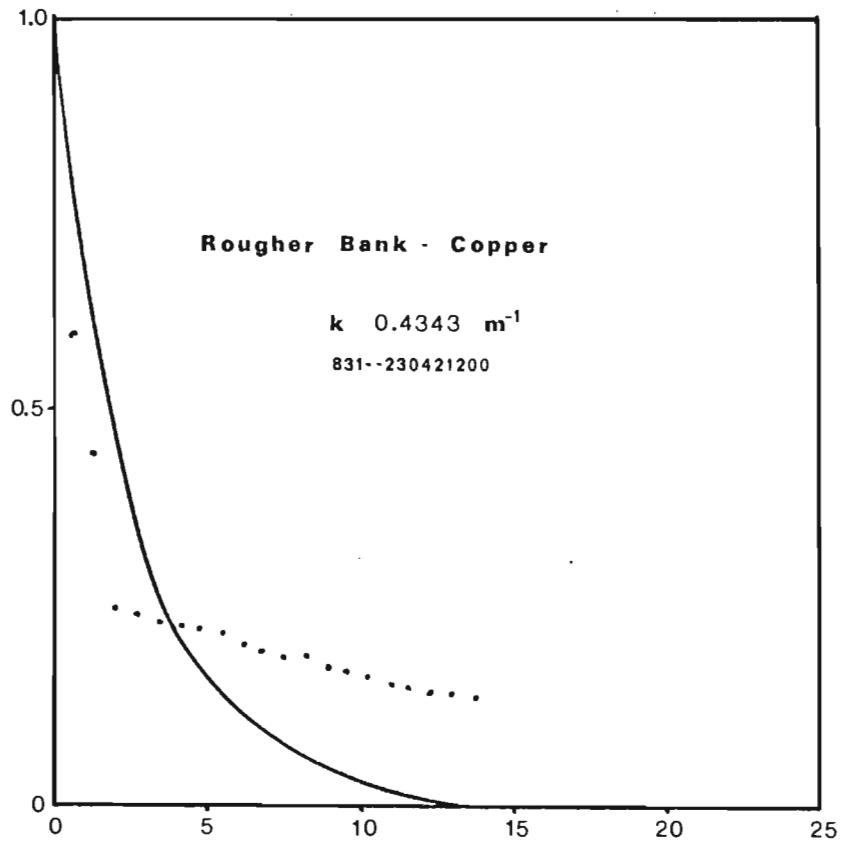
**13. Batch Tests - Single Rate Model**

**Fraction in Tailings vs Time - mins**



#### 14. Continuous Tests - Single Rate Model

Fraction in Tailings vs Mean Time - mins



**15. Continuous Tests - Single Rate Model**

**Fraction in Tailings vs Mean Time - mins**



### The Rate plus Non-floating Model

The rate plus non-floating first-order irreversible model analyzed at Rustenburg is a special case of the more general discretely distributed rate model proposed by Imaizumi and Inoue in 1965.<sup>31</sup> Difficulties with the single rate model gave rise to the understanding that flotation may well be better described by a series of rate constant values, each value governing the flotation of a fraction of the total mineralogical or metal species. Imaizumi and Inoue presented a graphical technique for estimating the first-order rate values and the fraction of material associated with that rate from a batch concentration versus time curve. Multiple linear regression may also be used to calculate the species 'k' distribution,<sup>32</sup> however, results vary with the discrete k values chosen and the method is highly sensitive to experimental error. The parameter estimation problems obviously become greater with increasing numbers of fractions and k values. The form of the discrete rate constant model applied at Rustenburg assumes that flotation of a species may be described by dividing the species into two fractions. One fraction has a rate constant of zero; it is non-floatable. The flotation of the remaining material may be described by a single first-order rate constant greater than zero.

Figures 16, 17 and 18 illustrate several fits of the rate plus non-floating model to the flotation of various species in a batch cell. In general the model describes batch flotation with reasonable accuracy, the non-floating fraction must be evident from the data however.

A number of continuous bank tests may also be used to test the flexibility of the rate plus non-floating model as shown in figures 19 and 20. The model structure in general is suitable but care must be exercised in using parameters regressed from

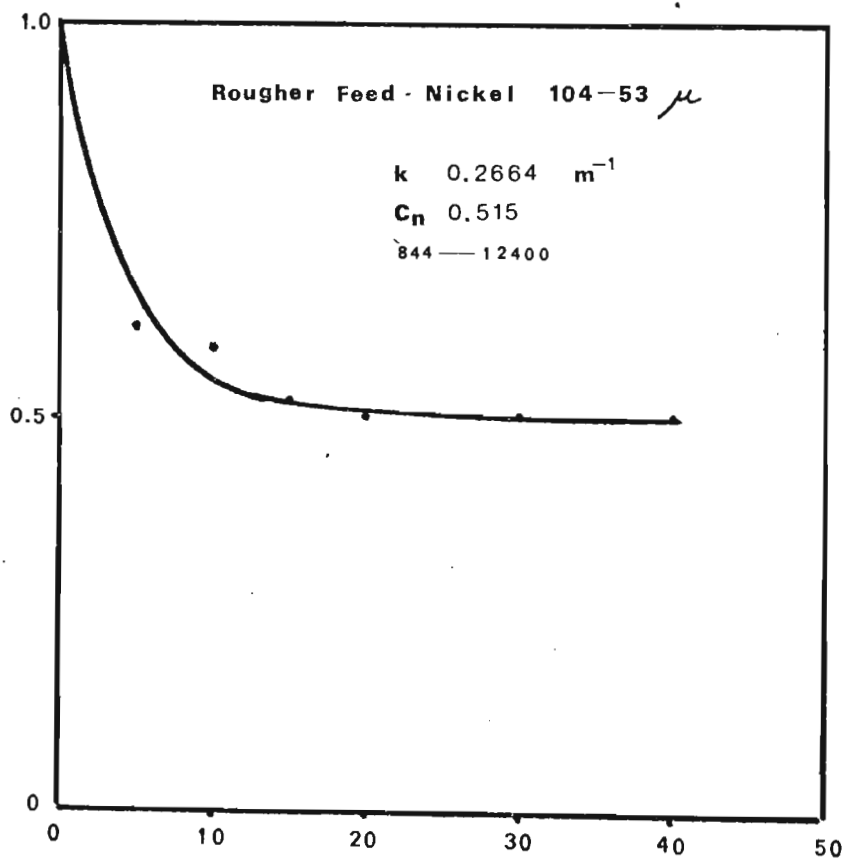
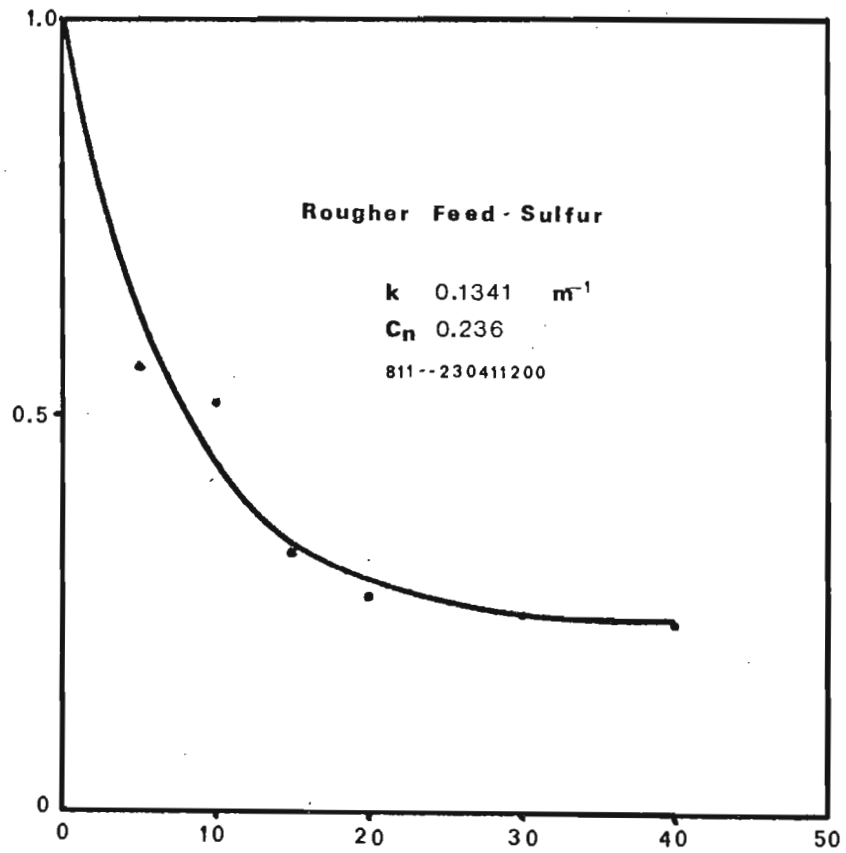
plant data. In an operating plant the flotation capacity is such that normal operation does not well define the fraction of a species with a rate constant of zero. Although the regression may fit the data well, no significance may be attached to the value of the non-floating fraction hence the consequent prediction of flotation recovery due to increases or decreases in plant capacity may be seriously in error.

A reasonable test of the rate plus non-floating model's predictive power may be analyzed through comparison of the operation of a single bank in the plant with the prediction based on kinetic parameters regressed from a batch test on the feed stream to that bank. The operation of a continuous bank may be predicted by the model through the residence time distribution. However, the model structure does not absorb any differences in machine characteristics such as impeller speed, bubble distribution, volume to surface ratios, etc. Figures 21, 22 and 23 illustrate the predictive capacity of the model and Table IV summarizes all the available data of this nature. Two problems are evident in predictions of this type; firstly, the froth removal rate during the batch test can greatly affect the accuracy of the prediction of continuous operation, this is particularly true of material which contains a high proportion of rapidly floating material as in the cleaner section at Rustenburg; secondly, the flotation of the total sulphide material appears to be equipment-specific in that predictions for the total sulphide material are seriously in error, on the other hand predictions for copper and nickel material are reasonably accurate. The sulphides at Rustenburg are primarily chalcopyrite, pentlandite and pyrrhotite; pyrrhotite is easily oxidized in the milling circuit thus its floatability may be more susceptible to agitation intensity and the scrubbing action in various types of flotation equipment and lead to the poor predictions for total sulphide material.

The assumption of the rate plus non-floating model that the first-order rate constants are time-invariant was tested by several pairs of batch tests on the rougher feed and rougher concentrate streams. In tests of this nature the froth was removed rapidly so as to minimize its effect on the kinetic parameters. For this testwork a sample of the feed to one of the four identical rougher banks was floated in a batch cell, immediately after this test was completed, a sample of the rougher concentrate stream (from all four rougher banks) was collected and floated in a batch cell. The regressed parameters of the rougher feed may be used to generate the theoretical parameters of the floated rougher concentrate material and this may be compared with the actual batch test flotation of the rougher concentrate stream. Figures 24 and 25 illustrate the ability of the model to predict the flotation of material in the rougher concentrate, theoretically this material will contain nothing of the non-floatable fraction in the rougher feed and will float at the same rate as the floatable fraction of the rougher feed. Table V contains a summary of test information for the rate plus non-floating model; the regressed values of the first-order rate constant and the fraction of non-floatable material are shown for both streams; also shown is the regressed rate constant for the concentrate stream where the non-floatable fraction is held at zero. Theoretically, these simple single rate constants of the rougher concentrate will equal the rate constant of the floatable fraction of the rougher feed. In general, the material floats at a faster rate in the rougher concentrate batch test than indicated by the rougher feed batch test, particularly so in the finer size fractions (less than 53 microns). As indicated in the table, the majority of tests were performed under normal plant conditions such that dextrin, a non-sulphide depressant, was added to the rougher concentrate before batch

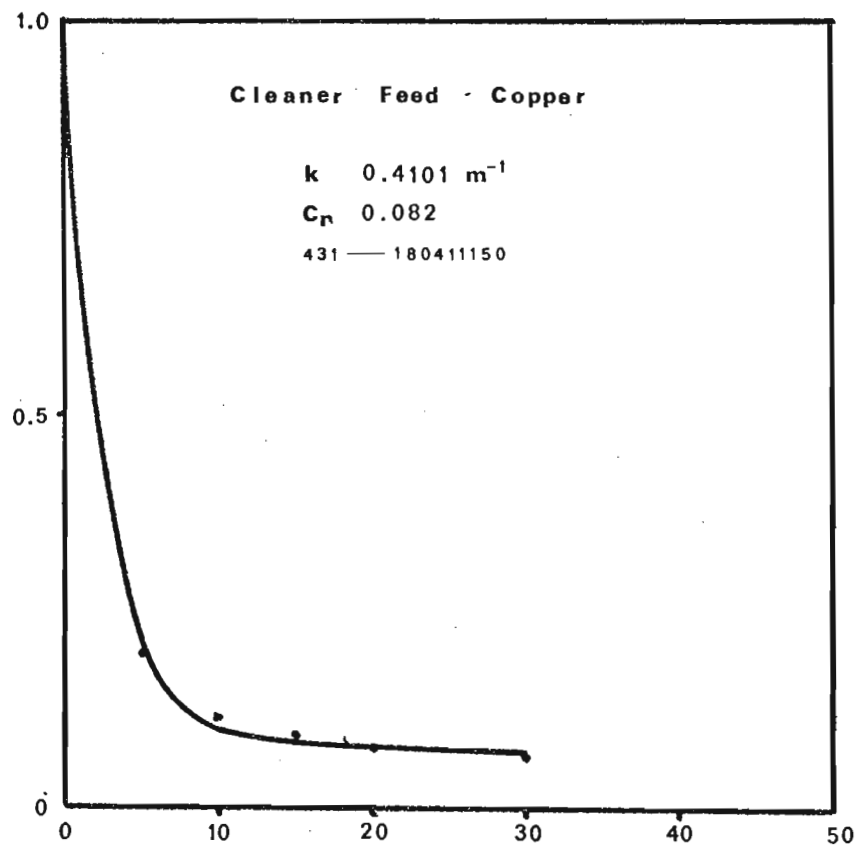
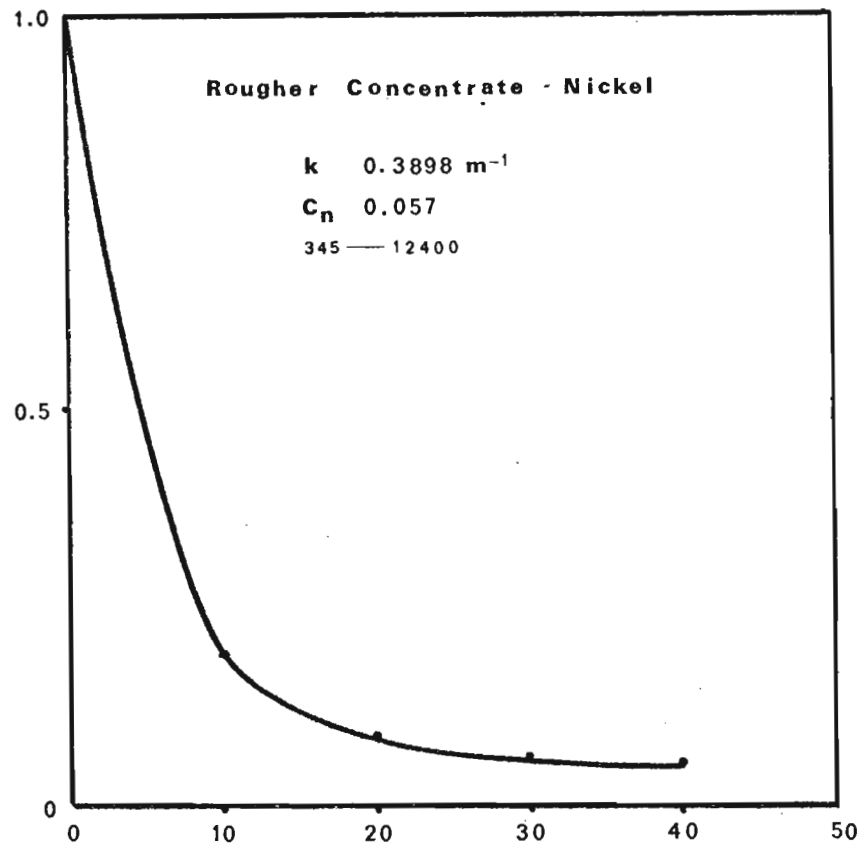


flotation. The one test performed without dextrin addition shows very similar results, however, and it must be concluded that dextrin addition does not significantly alter the floatability of the sulphide minerals.



16. Batch Tests - Discrete Model

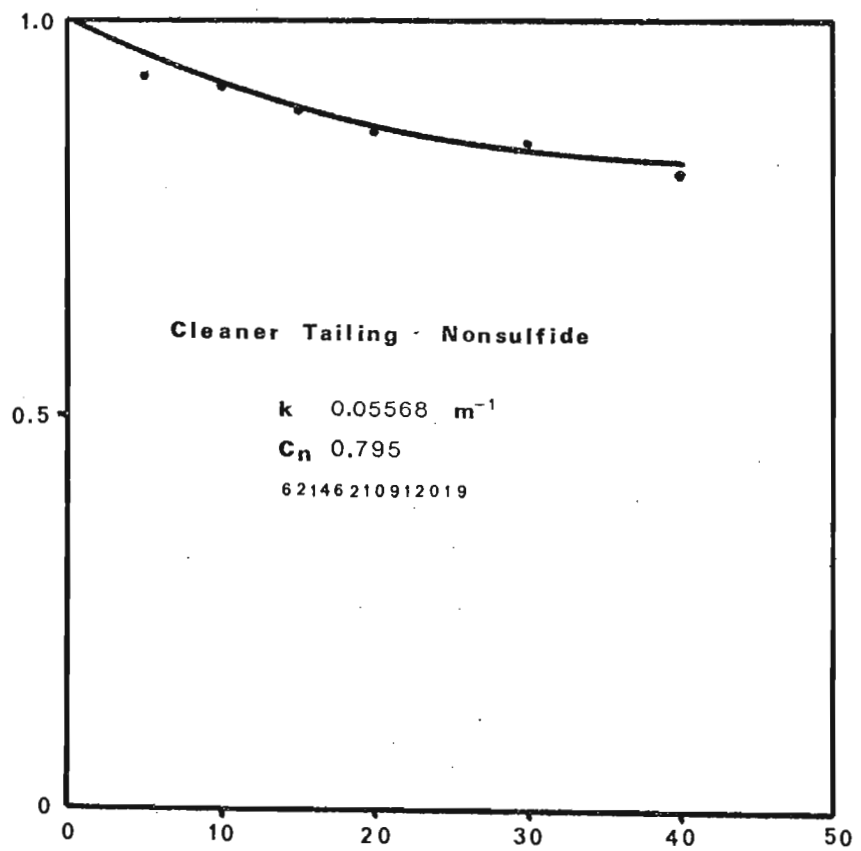
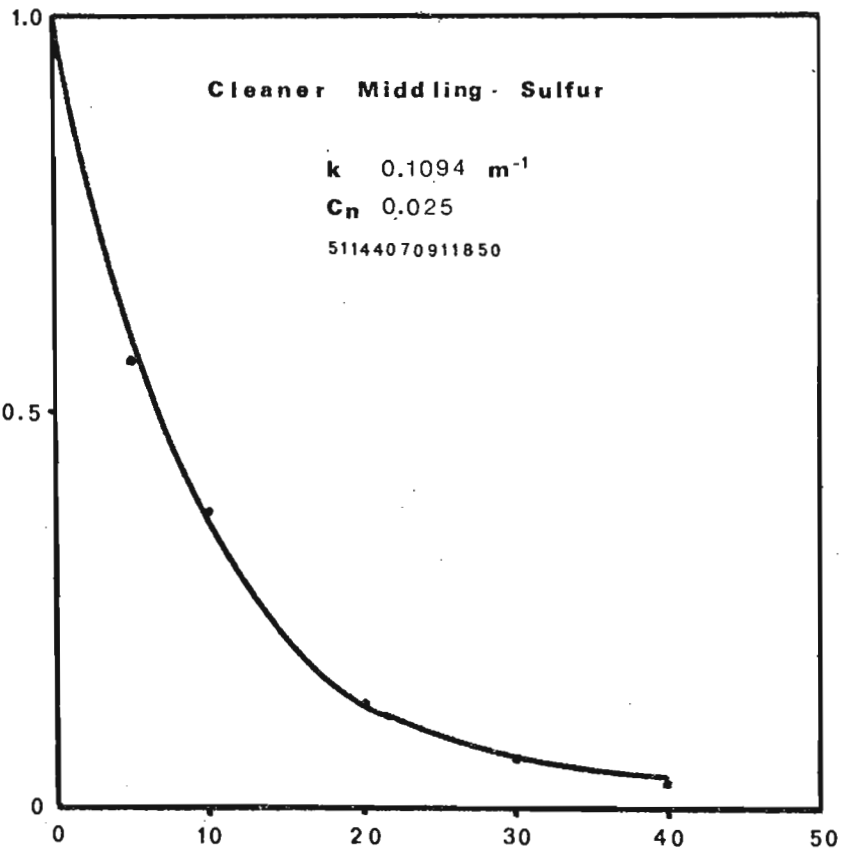
Fraction in Tailings vs Time - mins



17. Batch Tests - Discrete Model

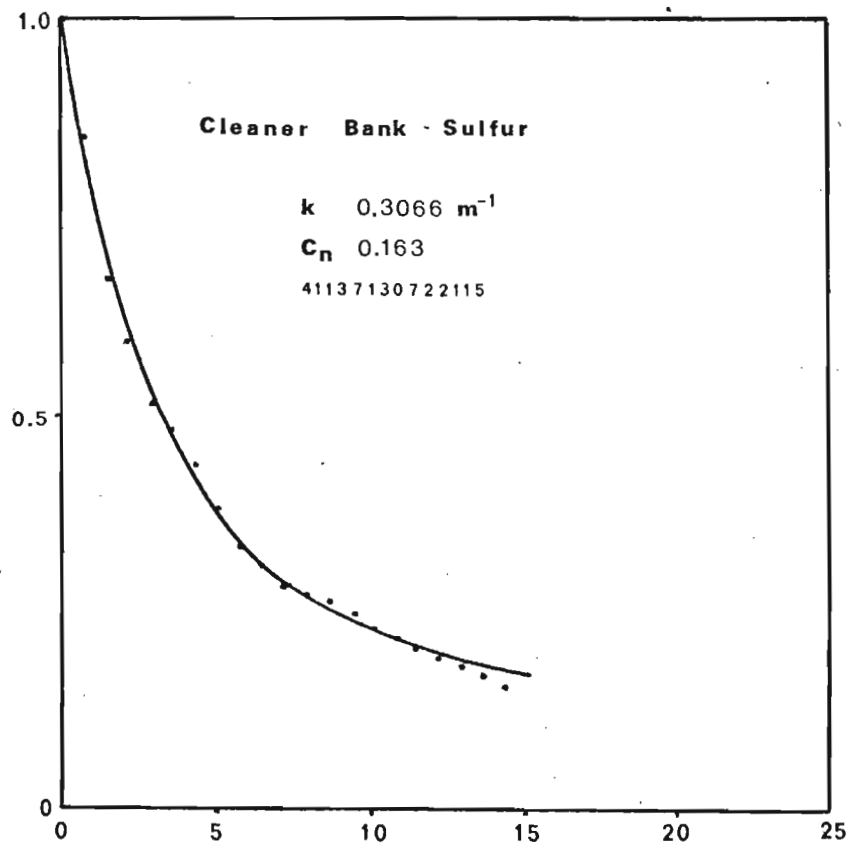
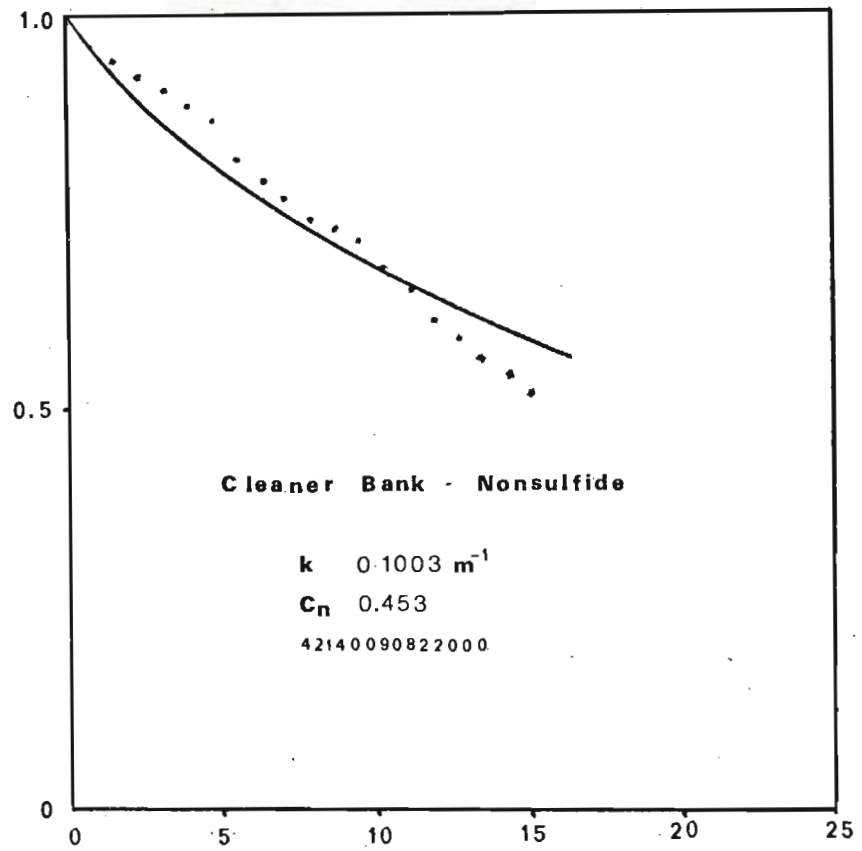
Fraction in Tailings vs Time - mins





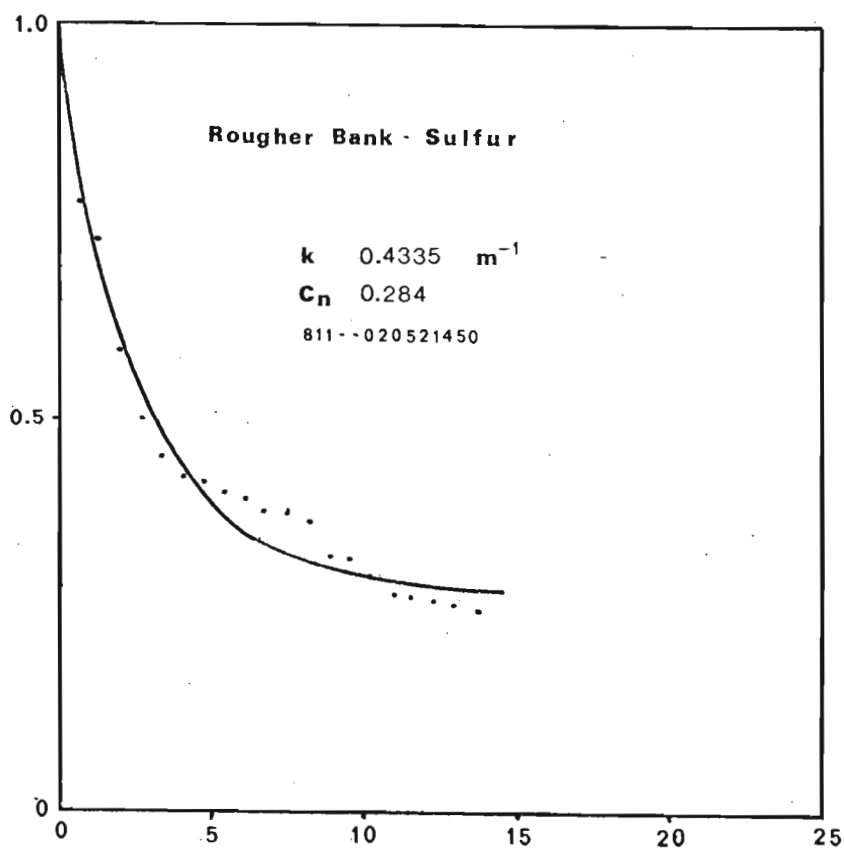
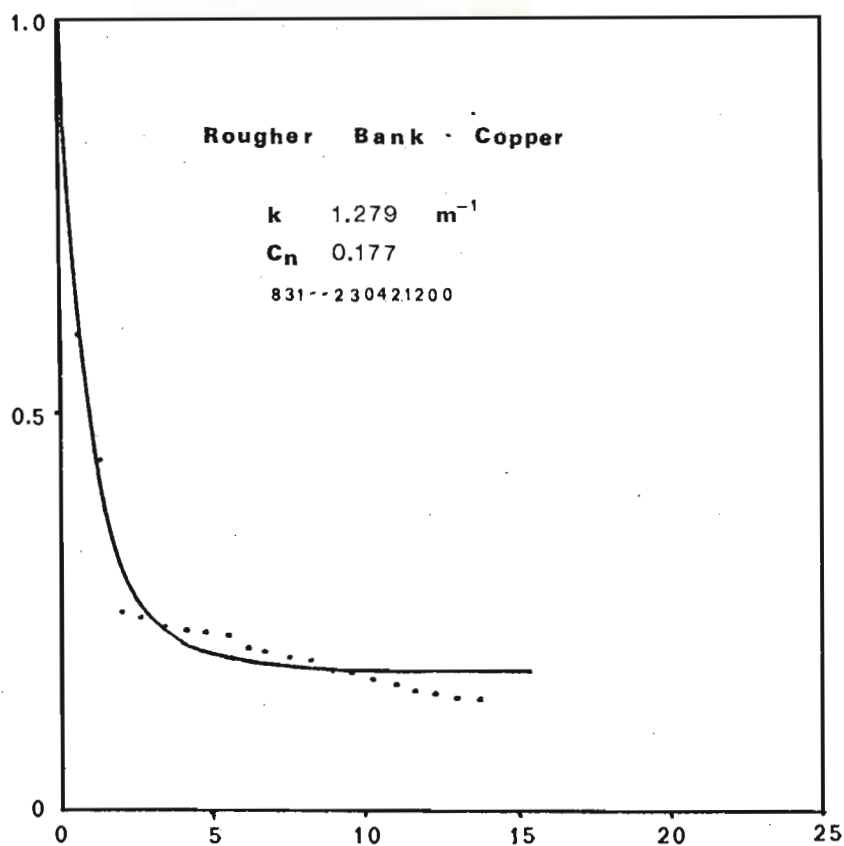
18. Batch Tests - Discrete Model

Fraction in Tailings vs Time - mins



19. Continuous Tests - Discrete Model

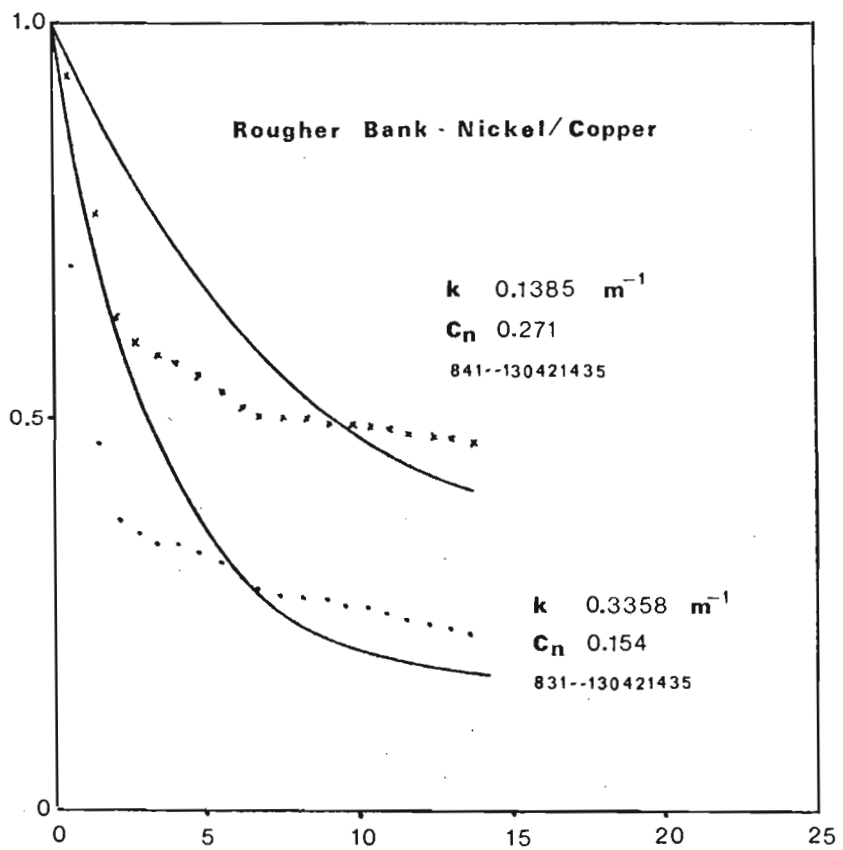
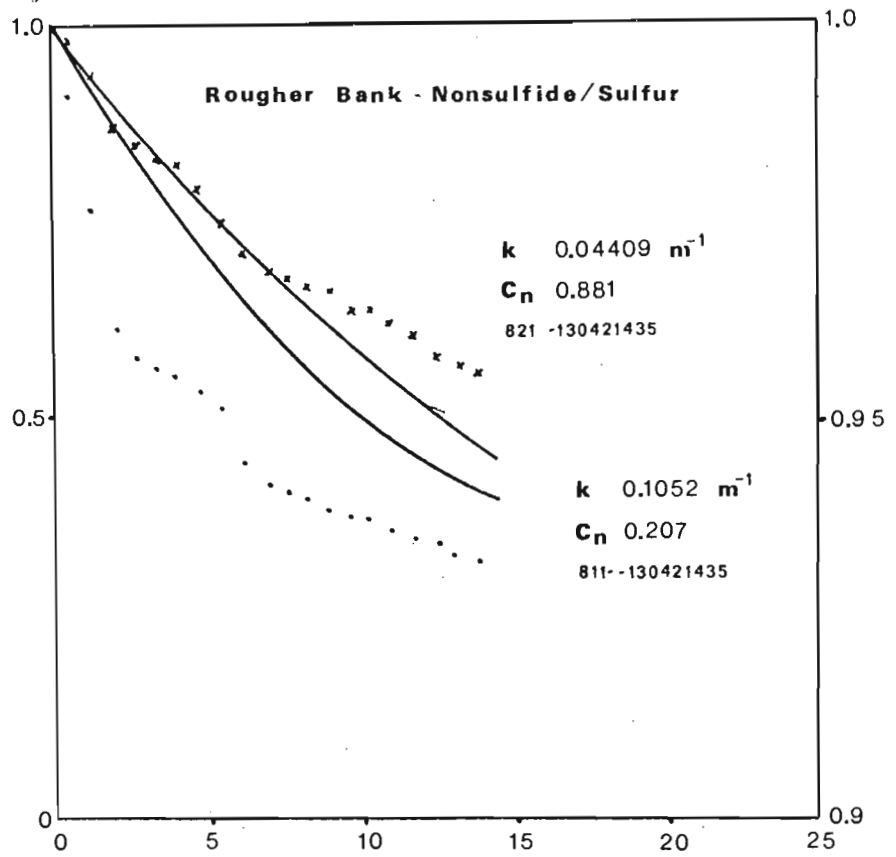
Fraction in Tailings vs Mean Time - mins



**20. Continuous Tests - Discrete Model**

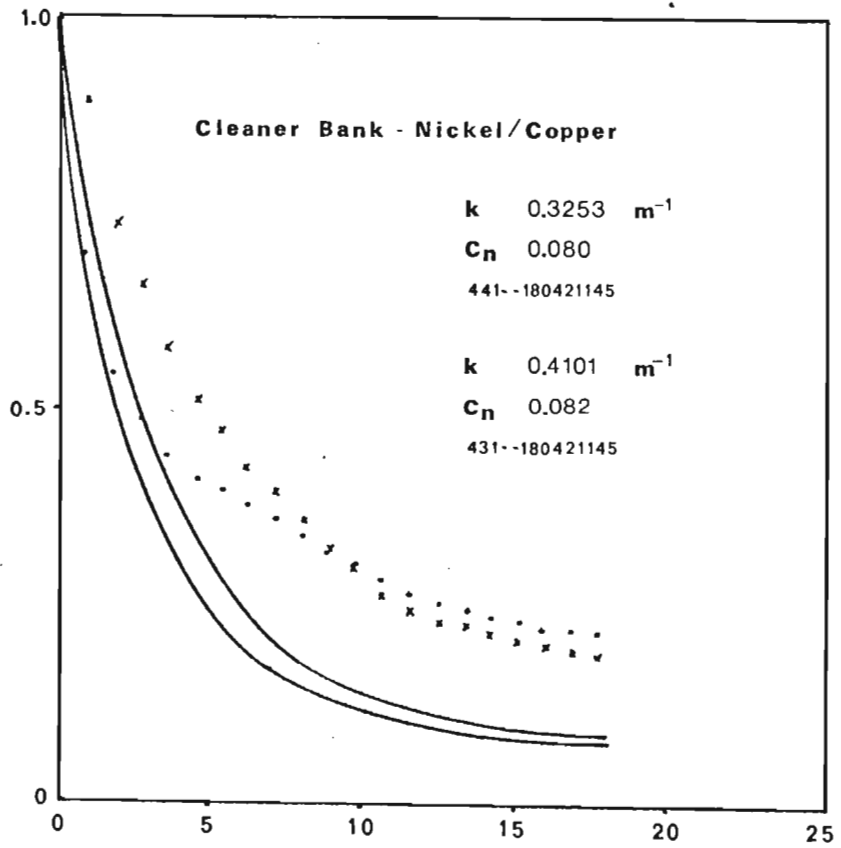
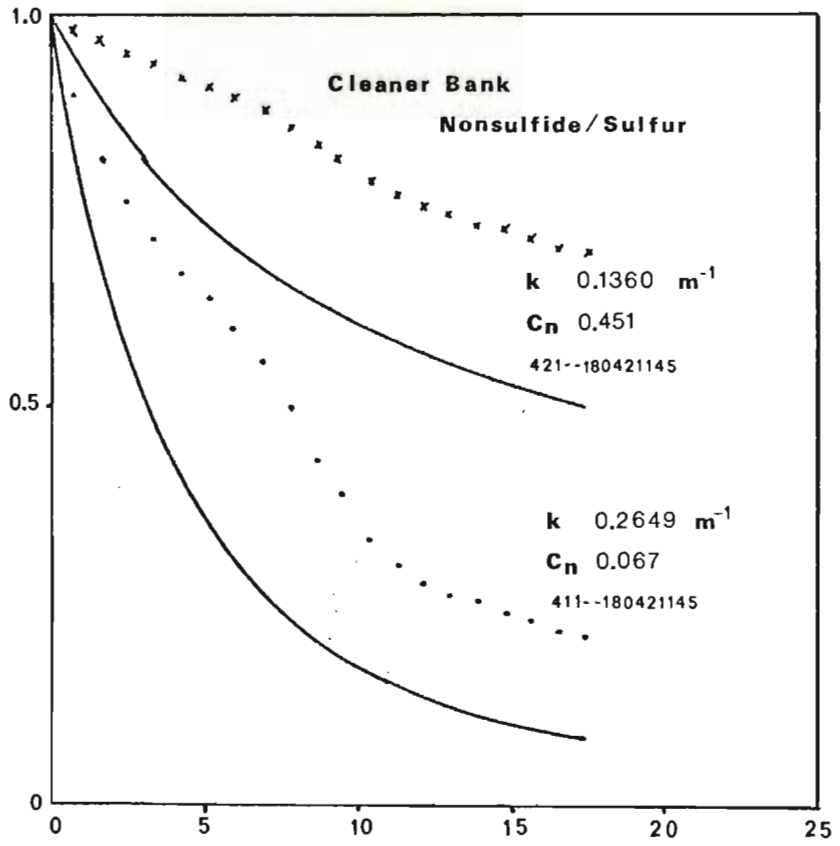
**Fraction in Tailings vs Mean Time - mins**





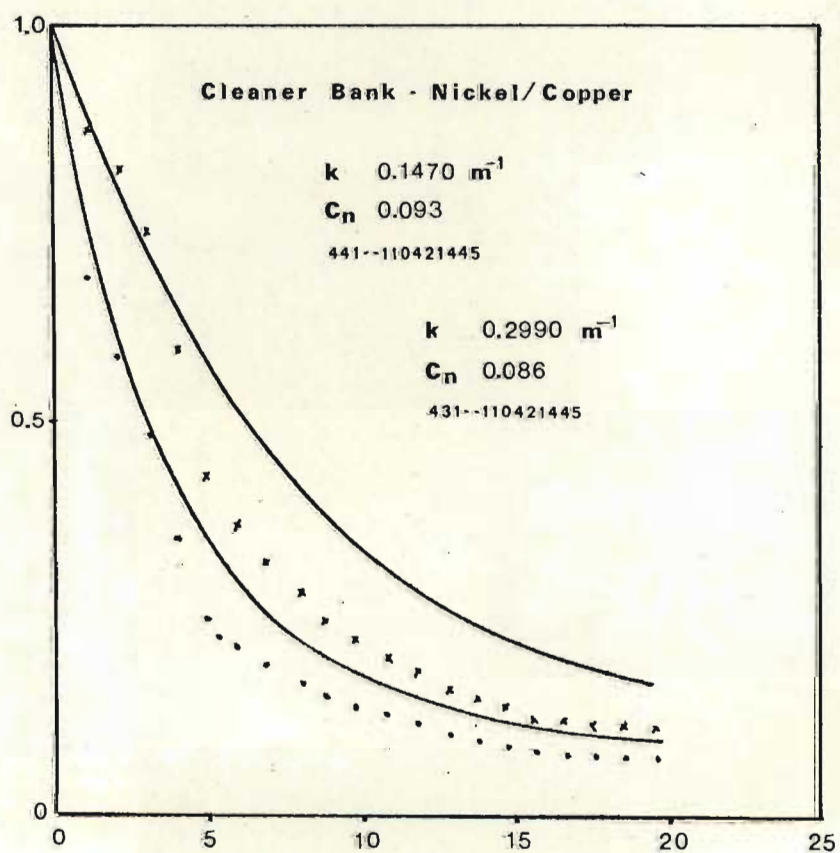
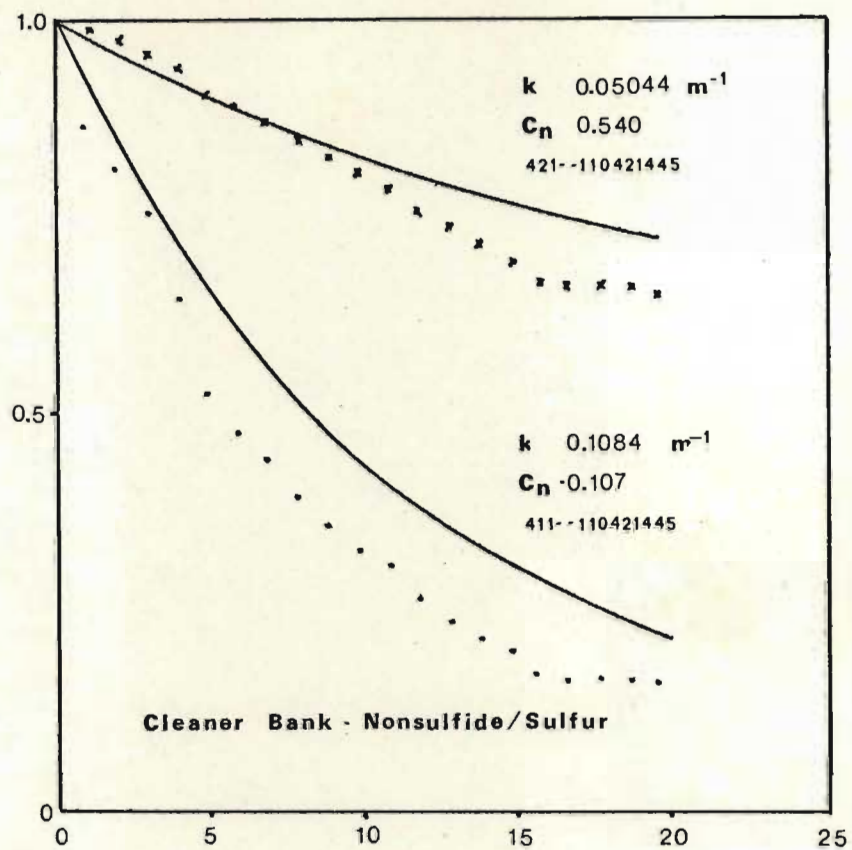
**21. Predicted Continuous Operation - Discrete Model**

**Fraction in Tailings vs Mean Time - mins**



## 22. Predicted Continuous Operation - Discrete Model

Fraction in Tailings vs Mean Time - mins



### 23. Predicted Continuous Operation - Discrete Model

Fraction in Tailings vs Mean Time - mins

TABLE IV. BATCH TEST PREDICTION OF CONTINUOUS OPERATION -  
RATE PLUS NON-FLOATING MODEL

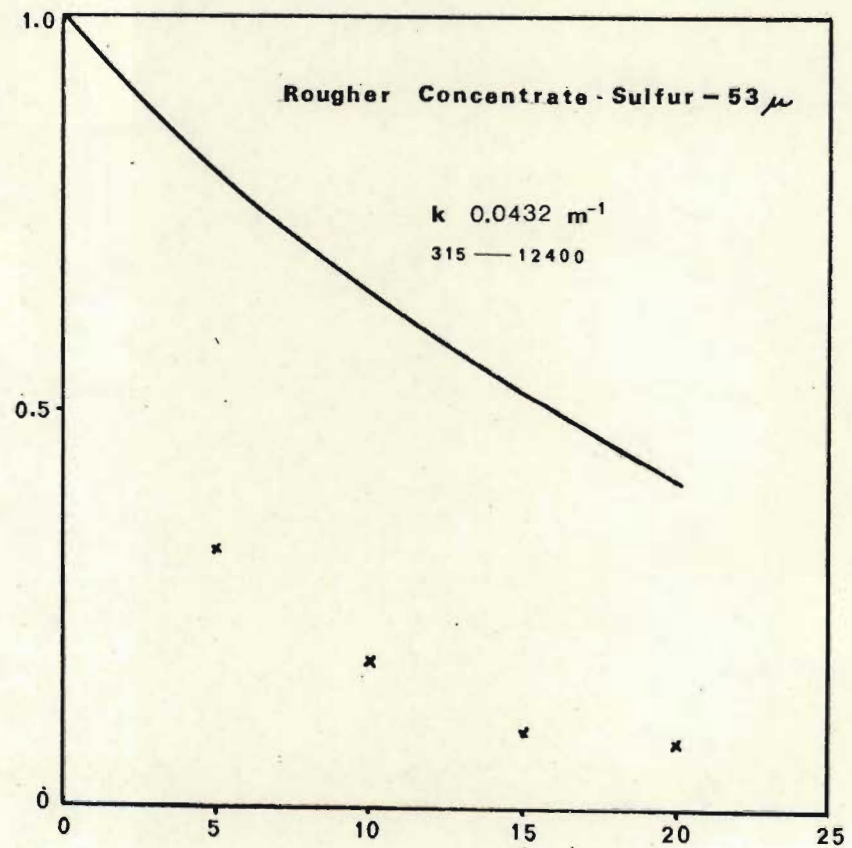
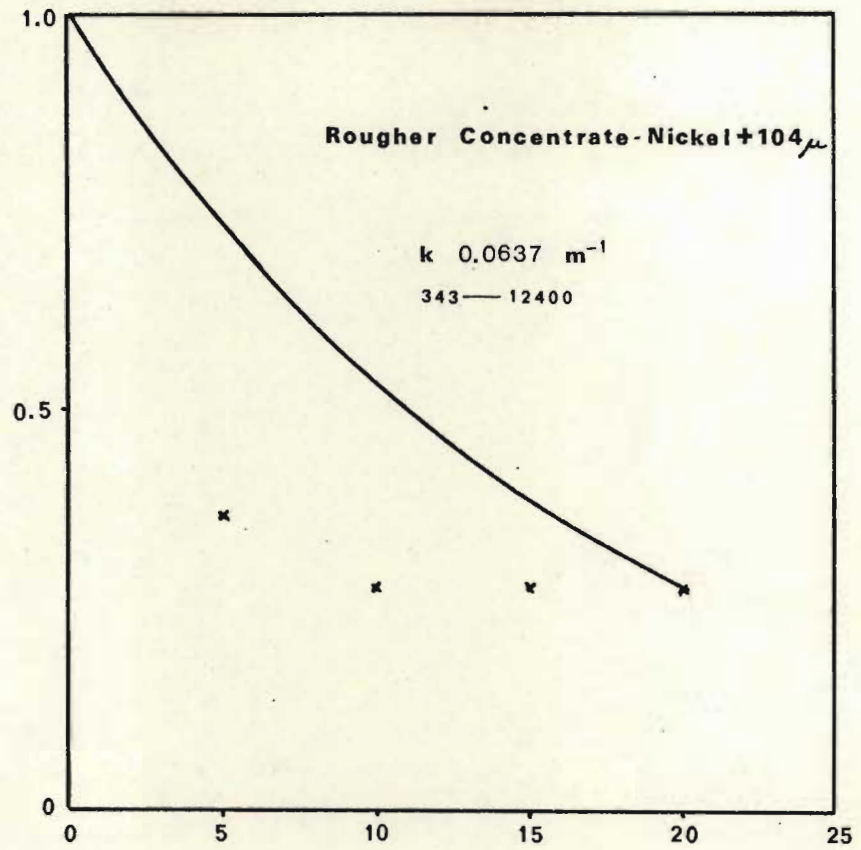
MATERIAL AND TEST	BATCH TEST CODE	CONTINUOUS TEST CODE	BATCH TEST PARAMETERS			TOTAL SQUARED ERROR IN CONCENTRATION PREDICTION
			Rate Constant $\text{min}^{-1}$	Non-floating Fraction	Sum of Squares	
Rougher Feed						
Sulphur	811--130411435	811--130421435	0,1052	0,207	0,01024	0,4635
Non-sulphide	821 "	821 "	0,04409	0,881	0,1763E-3	0,4173E-3
Copper	831 "	831 "	0,3358	0,154	0,5862E-2	0,2138
Nickel	841 "	841 "	0,1385	0,271	0,3432E-2	0,1773
Rougher Feed						
Sulphur	811--230411200	811--230421200	0,1341	0,236	0,01155	0,6742
Non-sulphide	821 "	821 "	0,05667	0,885	0,1274E-3	0,4503E-3
Copper	831 "	831 "	0,3452	0,179	0,6298E-2	0,5158
Nickel	841 "	841 "	0,2219	0,294	0,5120E-2	0,3615
Rougher Feed						
Sulphur	811--020511450	811--020521450	0,1147	0,292	0,9535E-2	1,188
Non-sulphide	821 "	821 "	0,05741	0,903	0,4222E-4	0,8002E-3



TABLE IV. (CONTINUED)

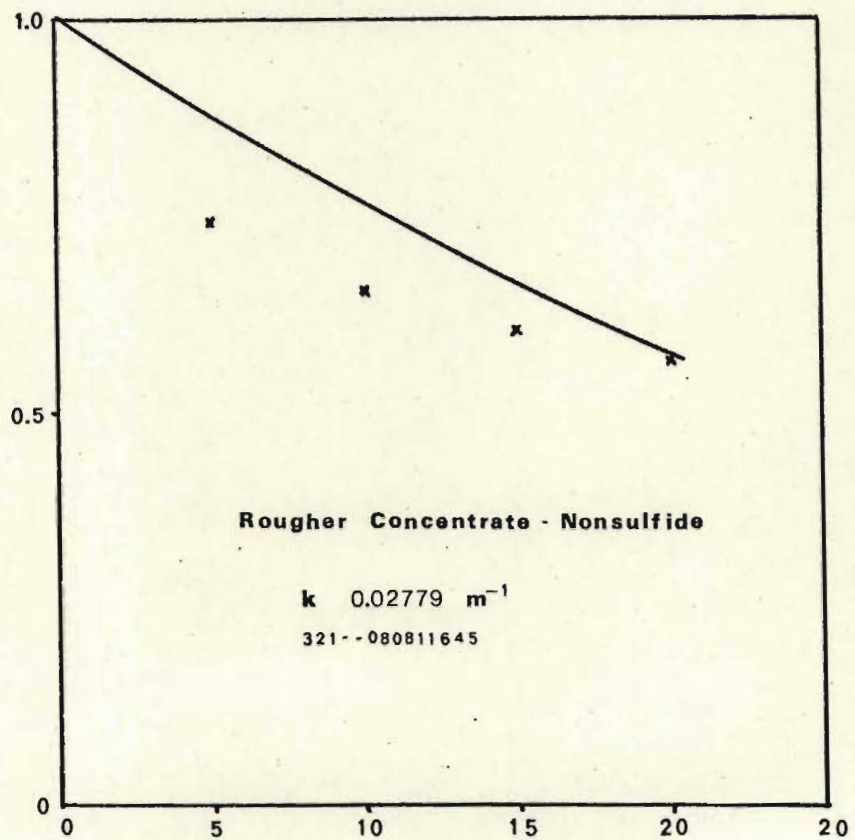
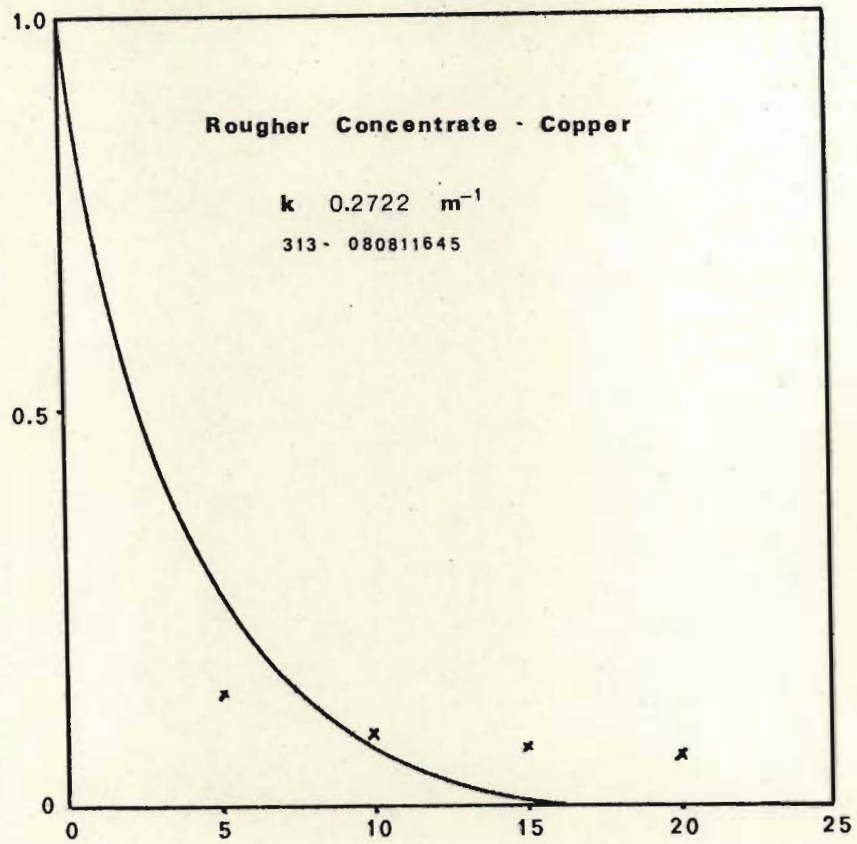
MATERIAL AND TEST	BATCH TEST CODE	CONTINUOUS TEST CODE	BATCH TEST PARAMETERS			TOTAL SQUARED ERROR IN CONCENTRATION PREDICTION
			Rate Constant min <sup>-1</sup>	Non-floating Fraction	Sum of Squares	
Cleaner Feed						
Sulphur	411--110411445	411--110421445	0,1084	0,107	0,2666E-2	0,1863
Non-sulphide	421     "	421     "	0,05044	0,540	0,1059E-3	0,04503
Copper	431     "	431     "	0,2990	0,086	0,4631E-2	0,04354
Nickel	441     "	441     "	0,1470	0,093	0,1261E-2	0,1518
Cleaner Feed *						
Sulphur	411--180411150	411--180421145	0,2649	0,067	0,1529E-3	0,9115
Non-sulphide	421     "	421     "	0,1360	0,451	0,6999E-4	0,6161
Copper	431     "	431     "	0,4101	0,082	0,6536E-3	0,4653
Nickel	441     "	441     "	0,3253	0,080	0,6576E-3	0,4212
Cleaner Feed						
Sulphur	411--300411430	411--300421430	0,1884	0,181	0,3180E-4	0,01434
Non-sulphide	421     "	421--     "	0,1081	0,566	0,2706E-4	0,8882E-2

\* This batch test was performed with a higher rate of froth removal than the others.



24. Predictions for Refloated Material - Discrete Model

Fraction in Tailings vs Time - mins



**25. Predictions for Refloated Material - Discrete Model**

**Fraction in Tailings vs Time - mins**



- 68 -  
TABLE V. PREDICTIONS FOR REFLOATED MATERIAL -  
RATE PLUS NON-FLOATING MODEL

Material	Rougher Feed Test Code	Rougher Feed Re- gressed Parameters		Rougher Concen- trate Test Code	Rougher Concentrate Regressed Parameters			Total Squared Error in Predicted Concentration
		Rate Constant min <sup>-1</sup>	Non- floating		Rate plus Non- Floating		Single Rate	
					Rate Constant (min <sup>-1</sup> )	Non- floating Fraction		
phur	811--080311935	0,1376	0,286	311--080312055	0,2353	0,084	0,1793	0,02660
	811--010311635	0,1968	0,362	311--010311800	0,4037	0,135	0,2211	0,03522
	811--220211920	0,1097	0,151	311--220212045	0,3628	0,097	0,2449	0,1644
	811--200211810	0,1524	0,208	311--200211945	0,3073	0,051	0,2554	0,06092
	811--140211805	0,1245	0,222	311--140211930	0,3668	0,074	0,2723	0,1396
phur (+208 microns)	812-----12400	0,1322	0,861	312-----12400	0,4759	0,398	0,0747	0,1967
(208-104 microns)	813       "	0,1238	0,597	313       "	0,2867	0,266	0,1058	0,05781
(104-53 microns)	814       "	0,1422	0,563	314       "	0,3215	0,145	0,1821	0,04375
( -53 microns )	815       "	0,0432	0,063	315       "	0,2543	0,083	0,1912	0,7461
per (208-104 microns)	833-----12400	0,2245	0,438	333-----12400	0,5969	0,118	0,3084	0,04532
(104-53 microns)	834       "	0,5051	0,213	334       "	0,6544	0,048	0,4723	0,6569E-2
( -53 microns )	835       "	0,05563	0,150	335       "	0,5938	0,029	0,4933	1,002
kel (208-104 microns)	843-----12400	0,2424	0,815	343-----12400	0,4108	0,274	0,1171	0,1772
(104-53 microns)	844       "	0,2664	0,515	344       "	0,3703	0,095	0,2517	0,01759
( -53 microns )	845       "	0,0637	0,232	345       "	0,3898	0,057	0,3073	0,6433
hur *	811--080811620	0,08672	0,123	311--080811645	0,2987	0,132	0,1798	0,1782
sulphide *	821       "	0,02779	0,861	321       "	0,1610	0,557	0,03482	0,02925
er *	831       "	0,2722	0,241	331       "	0,5176	0,072	0,3563	0,2049
el *	841       "	0,1318	0,299	341       "	0,3625	0,106	0,2339	0,08949

\* During these tests no dextrin was added to the rougher concentrate.



### The Gamma Function Model

The gamma function model is inherently better equipped to describe the flotation of a mineralogical or metal species in that its basic structure goes a step further than that of the discrete model and recognizes the individual nature of the flotation feed particles. The gamma function model assigns a continuous distribution of first-order rate constants to any species; the distribution takes the form of a modified gamma function and is completely defined by the mean ( $\bar{k}$ ) and variance ( $\sigma^2$ ) of the distribution.

$$W(k) = \frac{(\bar{k}/\sigma^2)^{\bar{k}^2/\sigma^2} (\bar{k}^2/\sigma^2 - 1) e^{-k\bar{k}/\sigma^2}}{\Gamma(\bar{k}^2/\sigma^2)}$$

Any number of algebraic continuous distributions of the first-order rate constants will accurately fit flotation batch test data, a bi modal distribution has been suggested by Harris and Chakavarti.<sup>33</sup> The advantage of the modified gamma function lies in that it can describe a wide variety of batch test curves with only two parameters defining the distribution; thus it is flexible and simple.

The gamma function model because of its continuous distribution of first-order rate constants, may be expected to accurately describe batch tests of any material and should perform as well if not better in predicting continuous plant operation from batch data kinetic parameters. A continuous distribution of rate constants may be expected to perform better in describing the refloatation of a species as well; however, like the other single phase models, the gamma function model will lack

the ability to handle the froth phase.

From the large number of batch tests performed, several species of flotation curves and the corresponding fits of the gamma function model are presented in Figures 26, 27 and 28. The gamma function model accurately describes batch flotation of any species varying from non-sulphide (very slow floating material) to copper (very fast floating material). Even when a large fraction of non-floating material is evident the model fit is accurate. The rate constant distribution commonly takes one of two forms shown in Figure 29. When  $\bar{k}^2/\sigma^2 - 1$  is negative, form A results and is indicative of a large fraction of non-floatable material. In the case of a species of this type it would be more theoretically correct to assign a non-floatable fraction to the species and describe the floatable fraction by the modified gamma function distribution. In example, for non-sulphide material it is obvious that some fraction is non-floatable and hence to describe the flotation of the whole non-sulphide species by a distribution of rate constants very near to zero is needlessly unrealistic.

Although the gamma function model describes continuous flotation by an integral without an analytical solution, i.e.

$$\alpha_j = \int_0^{\infty} \frac{W(k) dk}{(1+g)^j}$$

best fit parameters may be obtained by using Gauss-Laguerre quadrature methods and non-derivative search methods. Figures 30 and 31 illustrate the fit of the gamma function model to continuous data. The model form has some difficulty in fitting the combination of extremely fast and very slow floating material as in the case of copper material in the rougher feed;



other materials are handled accurately however.

As with the discrete model, the gamma function model may be used to predict the operation of a continuous flotation bank from the kinetic parameters regressed from a batch test of the feed stream to that bank. Figures 32, 33 and 34 illustrate the predictive capacity of the model and Table VI summarizes all of the data of this nature. Again as with the discrete model, the problem of froth removal and the fact that the rate constants are equipment-specific, are both evident. In the majority of cases, however, the gamma function model more closely describes continuous flotation than does the rate plus non-floating model. It is more accurate simply because it assigns an entire distribution of rates to the flotation feed species.

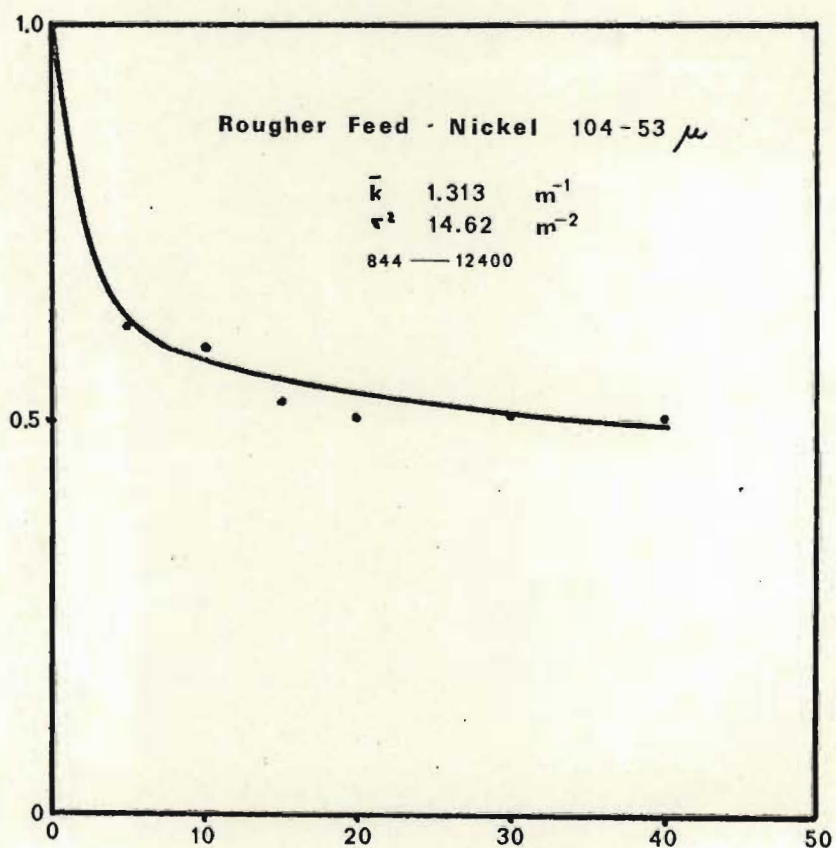
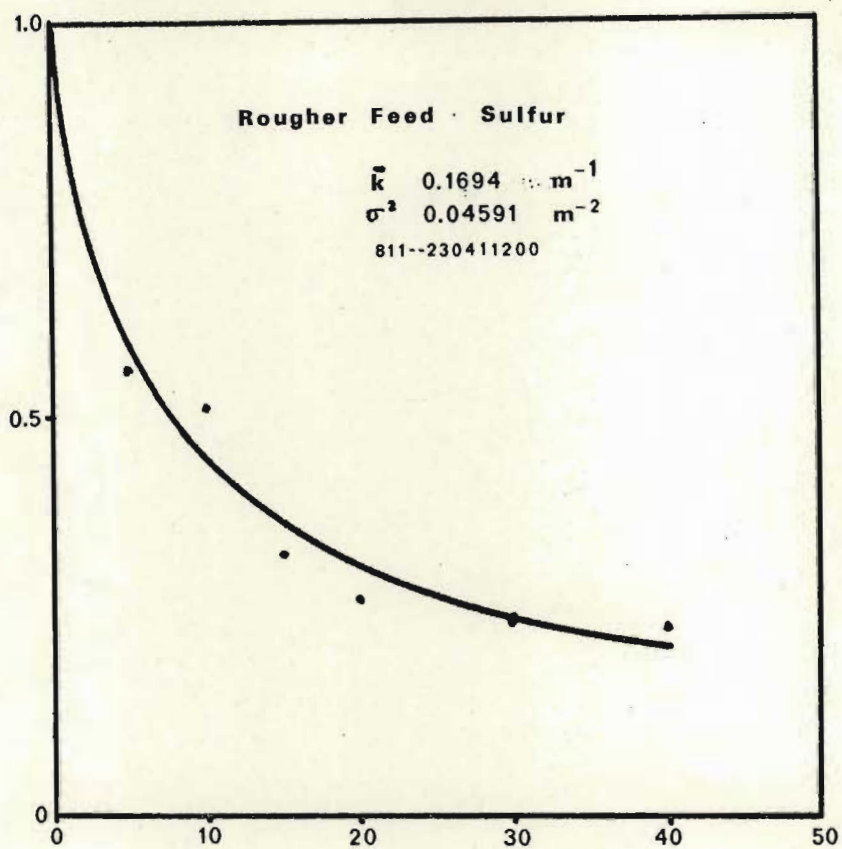
The applicability of the gamma function model must also be tested as with the rate plus non-floating model, with regard to the prediction of the reflation of concentrate streams. The batch tests performed on the rougher feed stream and rougher concentrate stream may be analyzed. Given the kinetic parameters  $\bar{k}$  and  $\sigma^2$  regressed from the rougher feed batch test, the kinetic parameters for the plant rougher concentrate may be calculated.

$$\bar{K} = \frac{\int_0^{\infty} k \underline{r}^T \underline{g} \underline{C}^{-1} \underline{\epsilon} W(k) dk}{\int_0^{\infty} \underline{r}^T \underline{g} \underline{C}^{-1} \underline{\epsilon} W(k) dk}$$

$$\sigma^2 = \frac{\int_0^{\infty} (k - \bar{K})^2 \underline{r}^T \underline{g} \underline{C}^{-1} \underline{\epsilon} W(k) dk}{\int_0^{\infty} \underline{r}^T \underline{g} \underline{C}^{-1} \underline{\epsilon} W(k) dk}$$

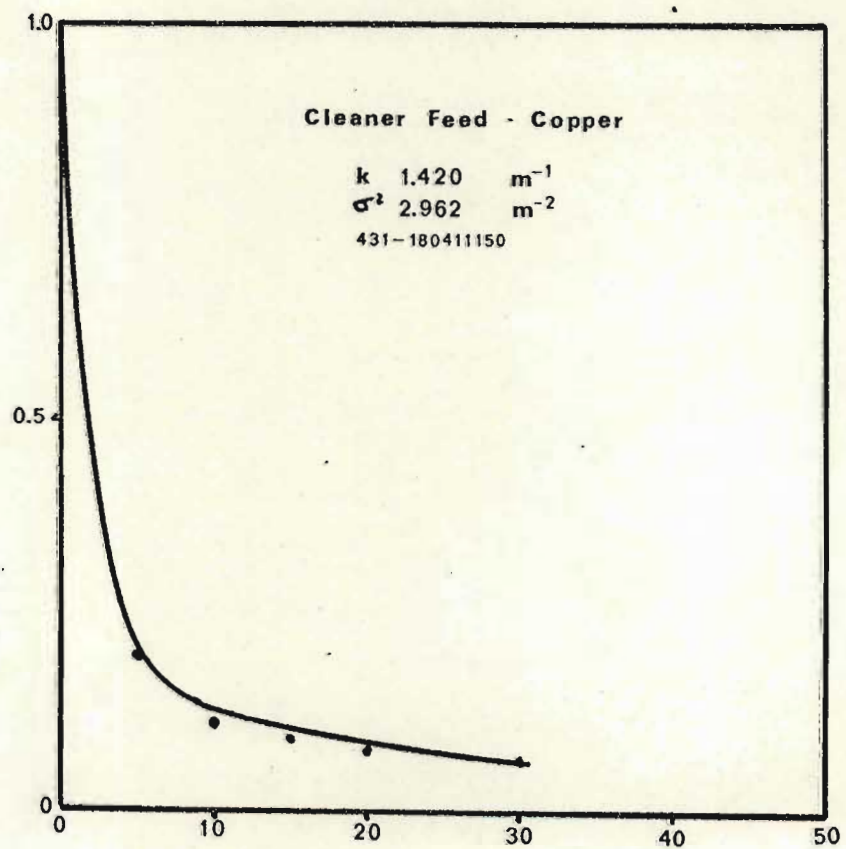
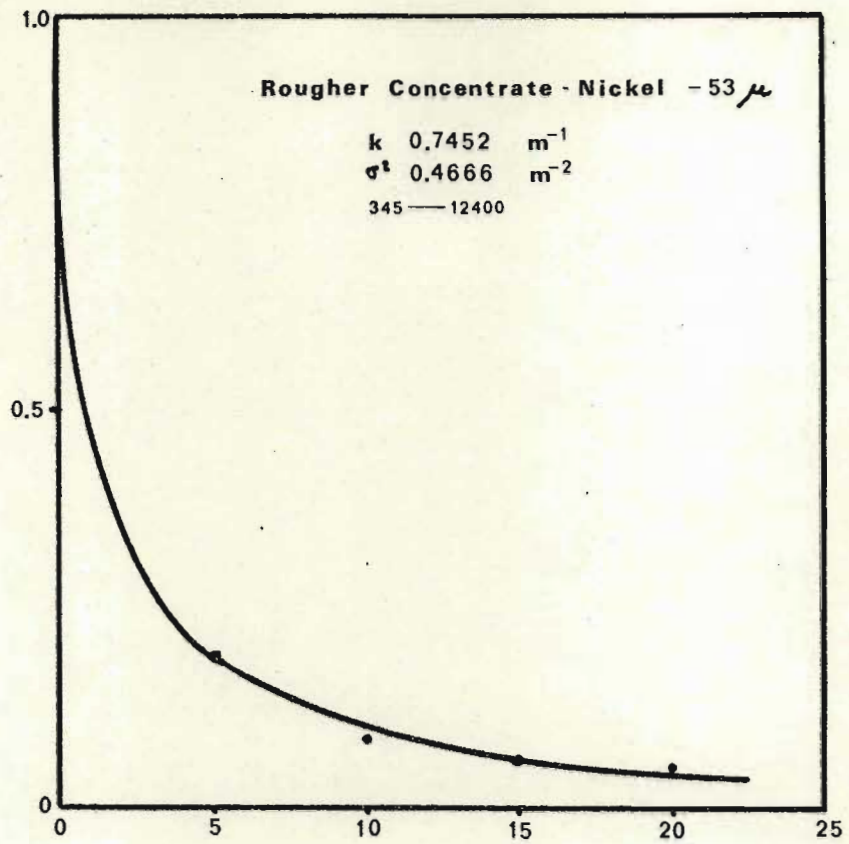
Once the kinetic parameters of the rougher concentrate stream have been calculated a predicted batch float of the rougher concentrate may be compared to the actual test results. Figures 35 and 36 illustrate the accuracy of the gamma function model in this capacity and Tables VII & VIII summarize all of the available data. In Table VIII, three parameters were used to describe the rougher feed test, two rate constant distribution parameters for the floatable material and a non-floating fraction. The addition of a third parameter here changes the calculated rate constant distribution in the rougher concentrate and improves the predictive abilities of the model, thus it may be of value to consider the gamma function rate constant distribution as descriptive of only the floatable material in a species. Again in the majority of cases, the gamma function model performs more accurately than the rate plus non-floating model. Due to the results of previous sections it might be expected that the accuracy for total sulphide material would be worse than that of copper and nickel, generally speaking this is the case. In total, predictions are still poor and especially so for material smaller than 53 microns.





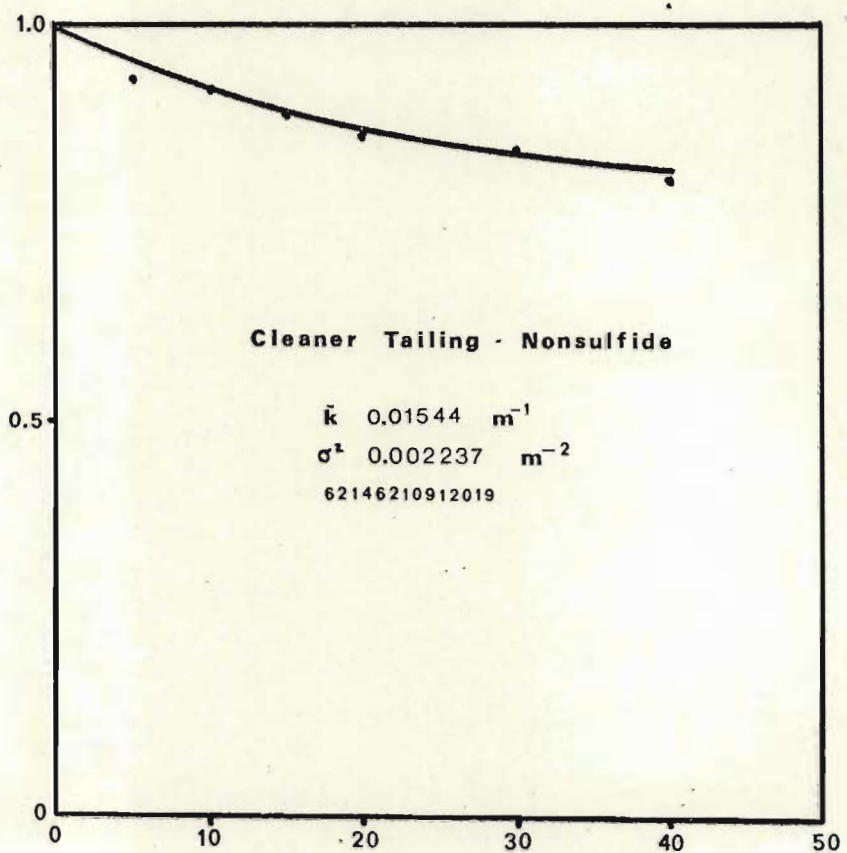
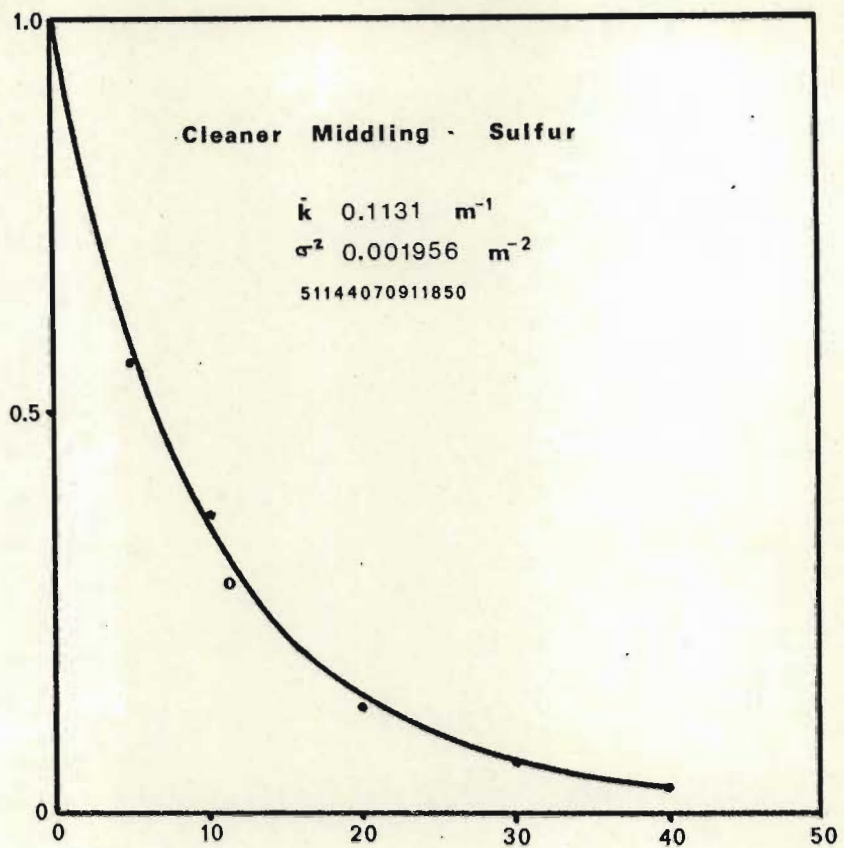
26. Batch Tests - Gamma Function Model

Fraction in Tailings vs. Time - mins



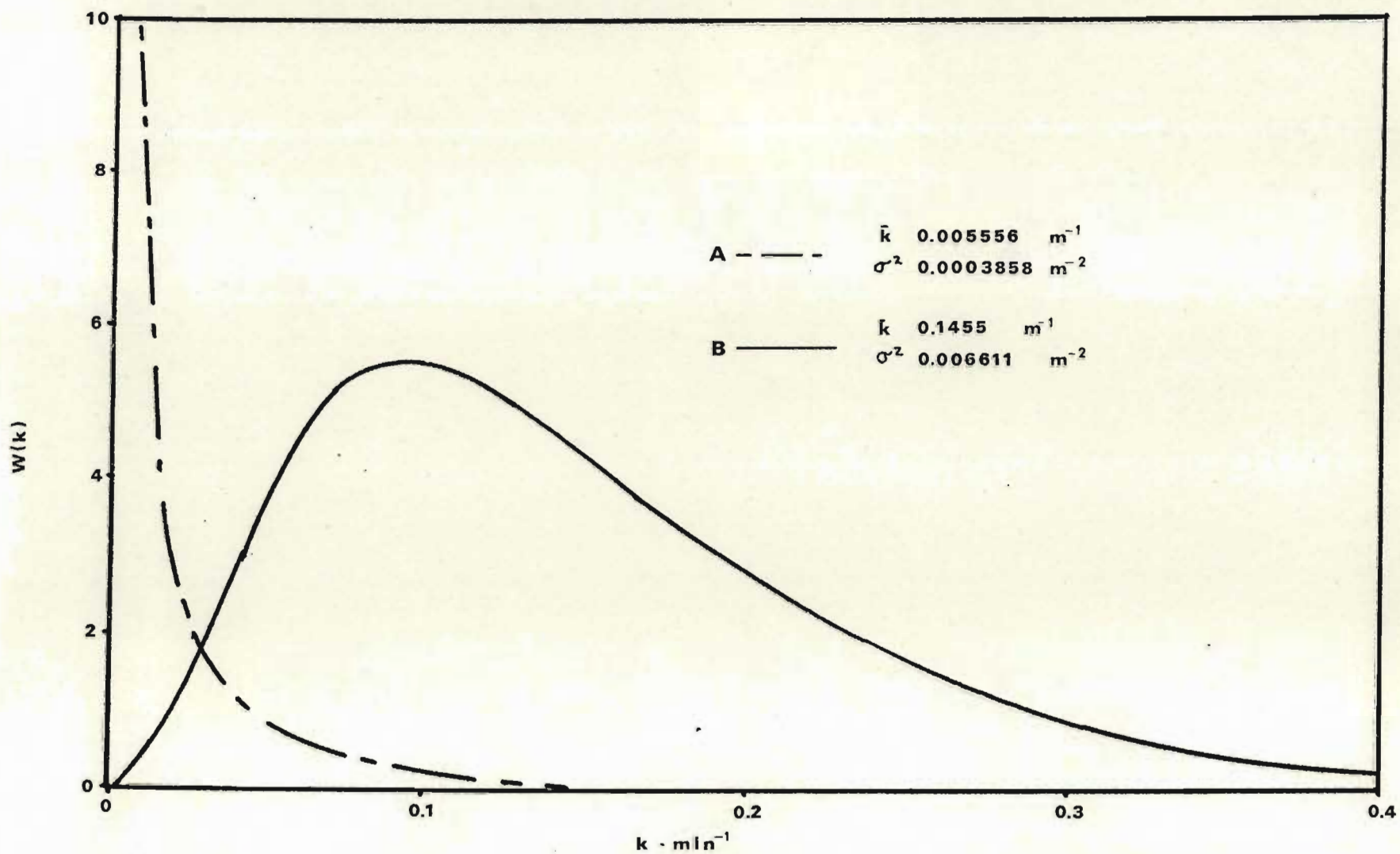
27. Batch Tests - Gamma Function Model

Fraction in Tailings vs Time - mins



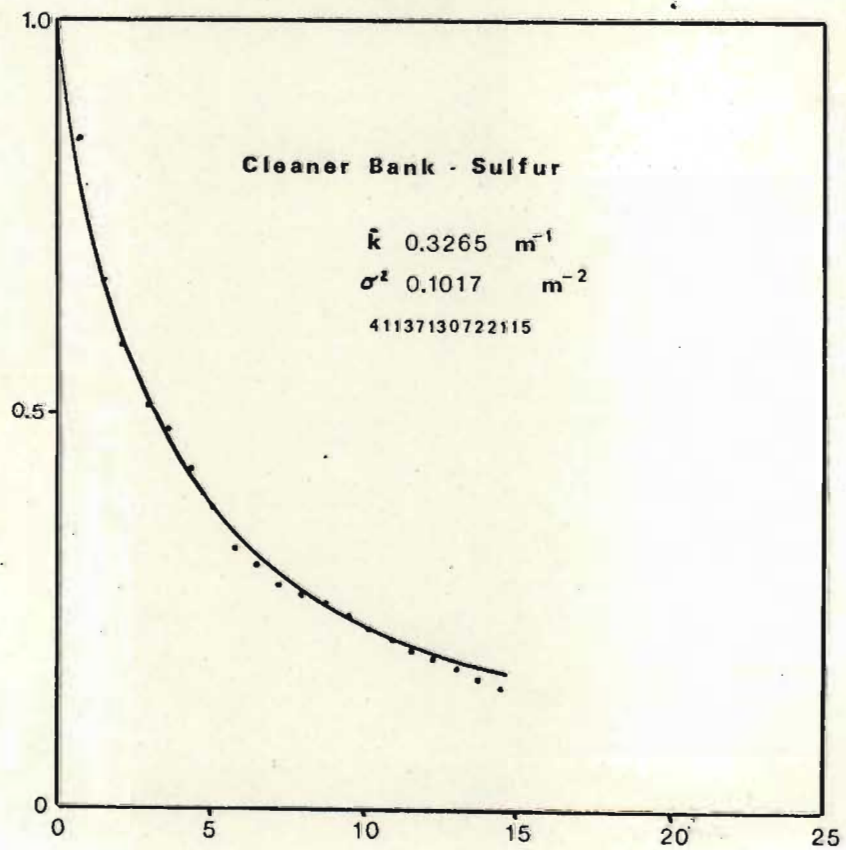
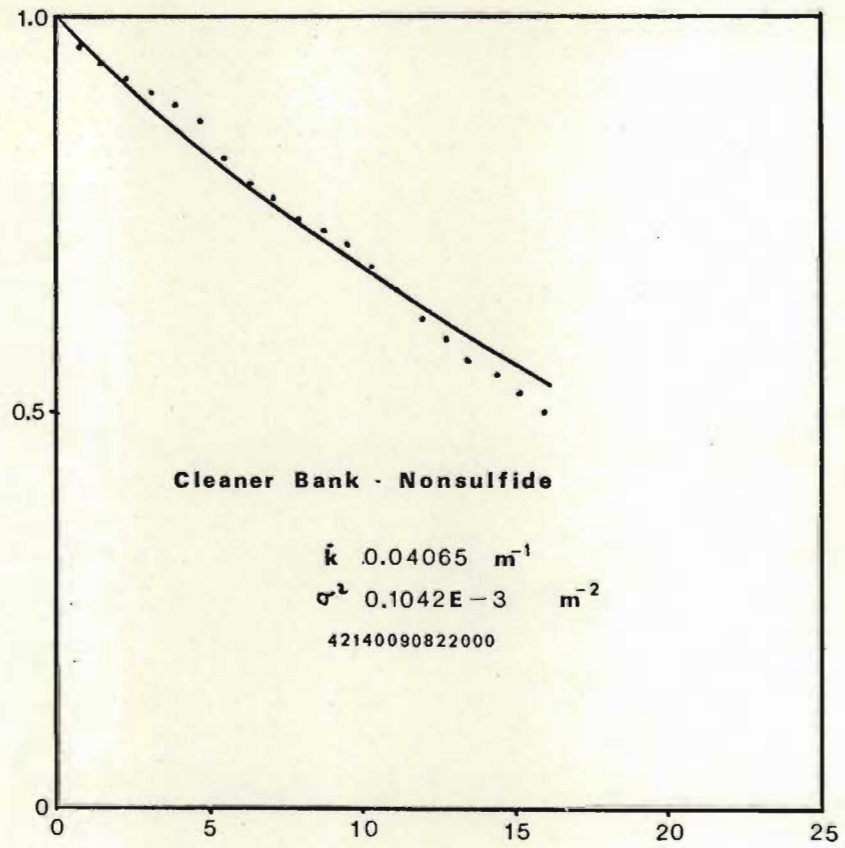
28. Batch Tests - Gamma Function Model

Fraction in Tailings vs Time - mins



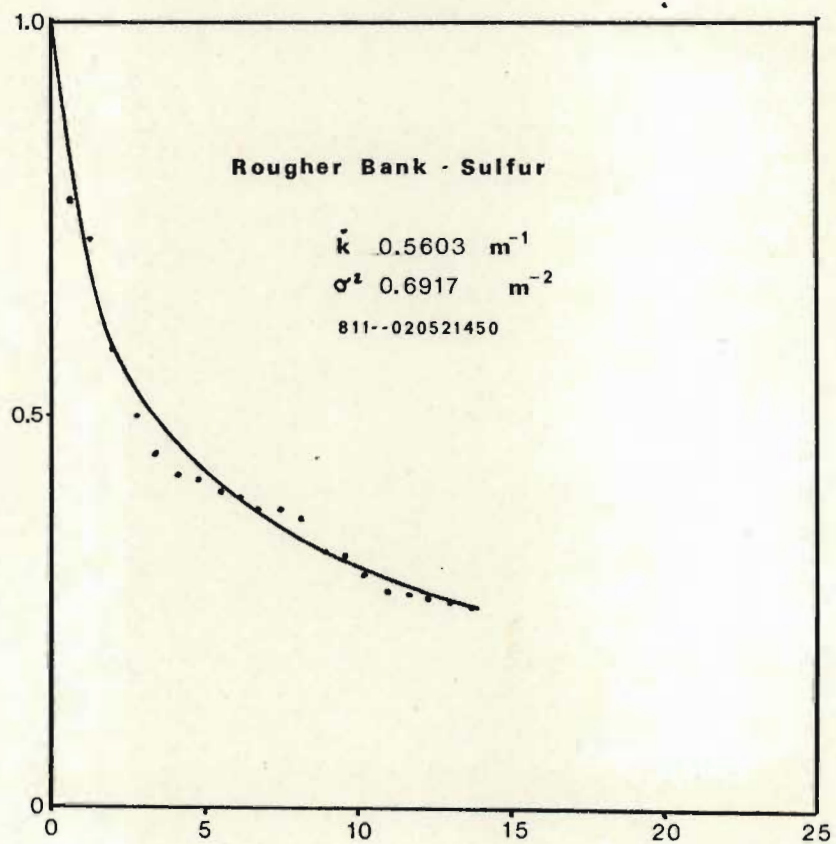
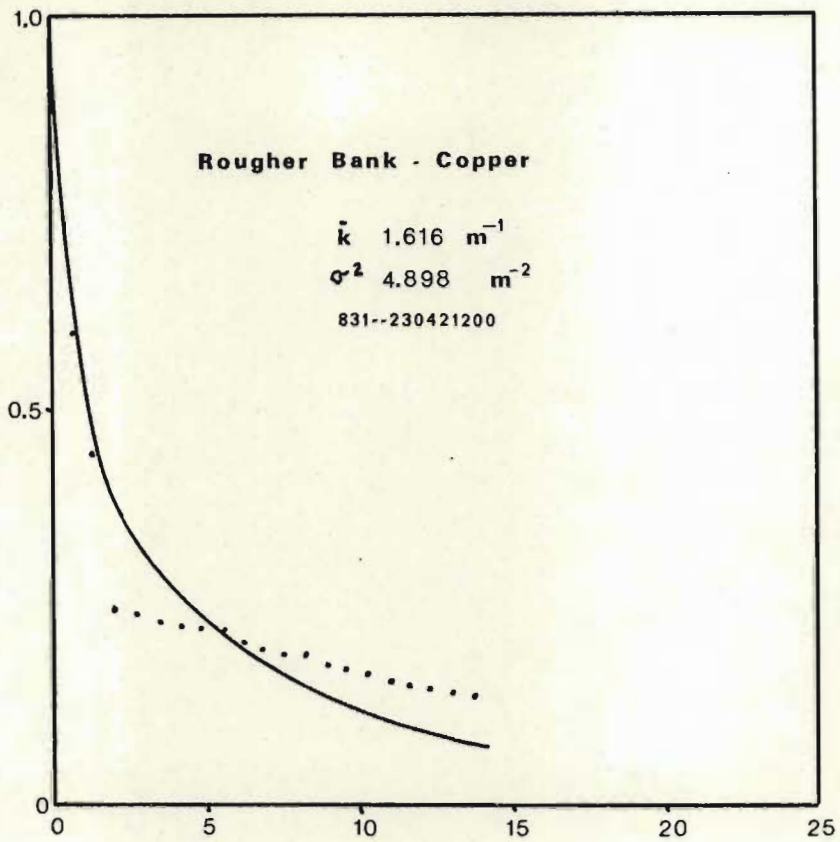
29. Typical Rate Distribution





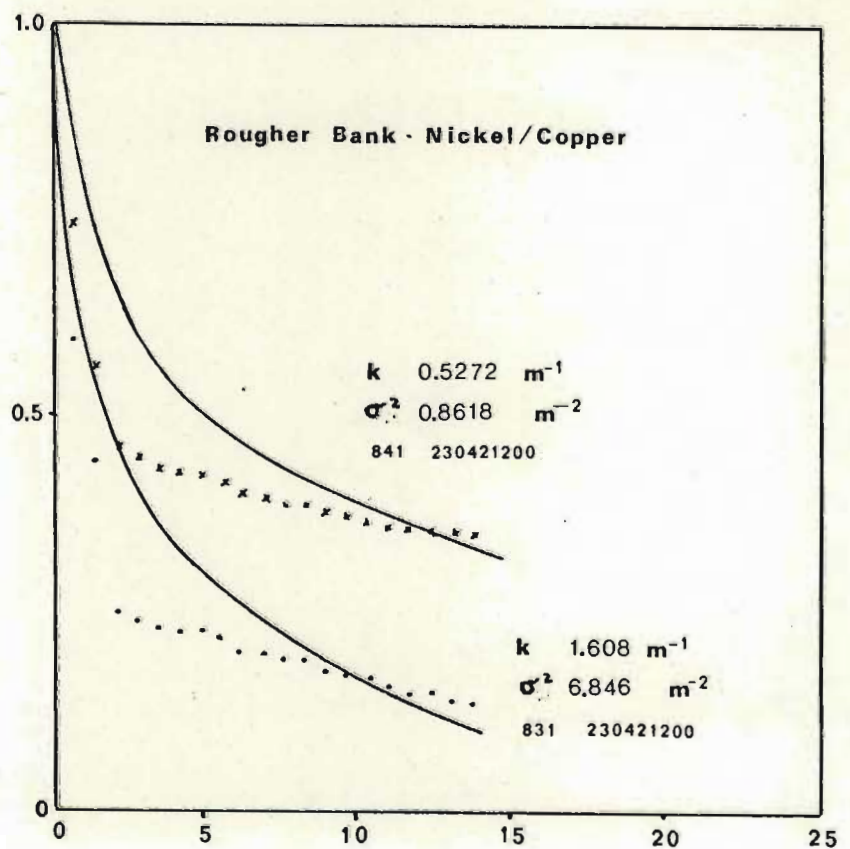
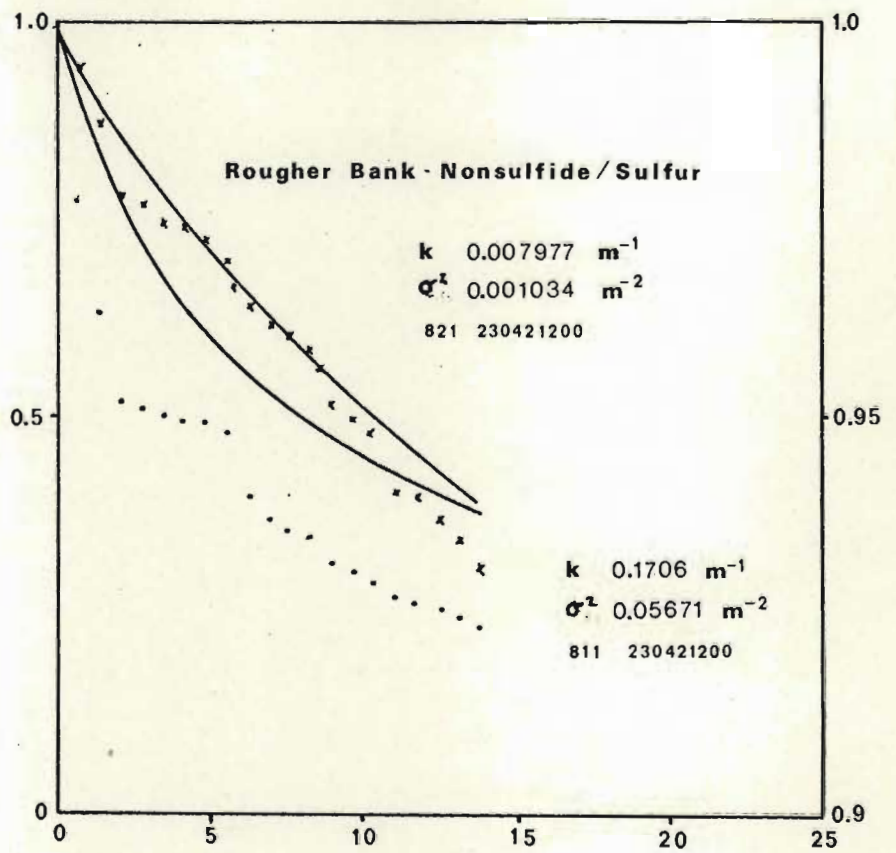
### 30. Continuous Tests - Gamma Function Model

Fraction in Tailings vs Mean Time - mins



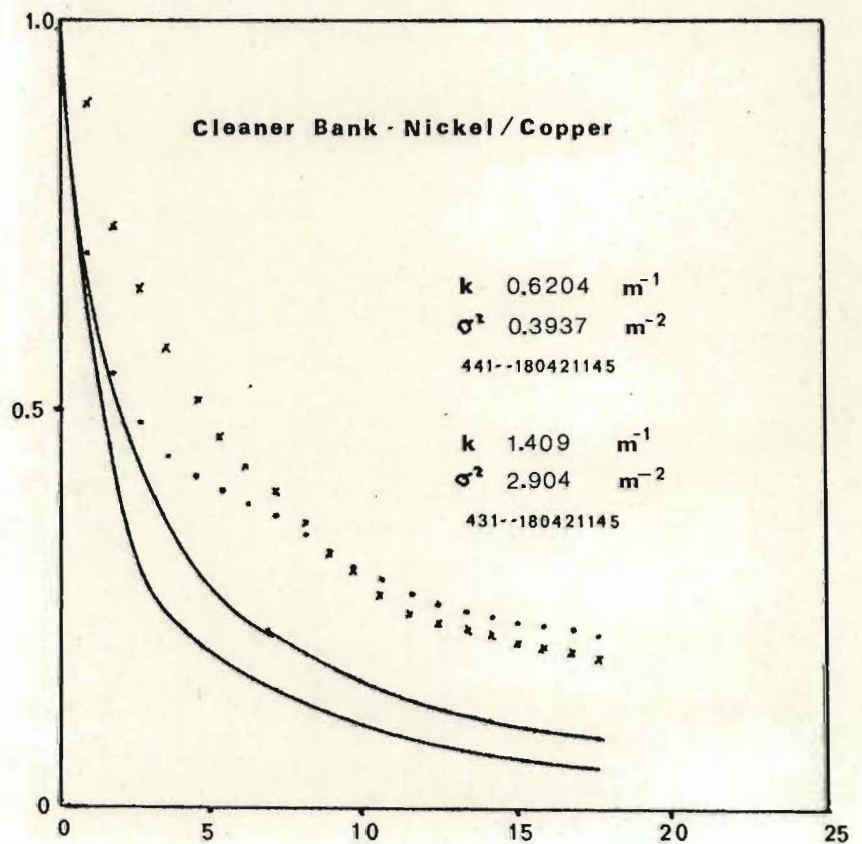
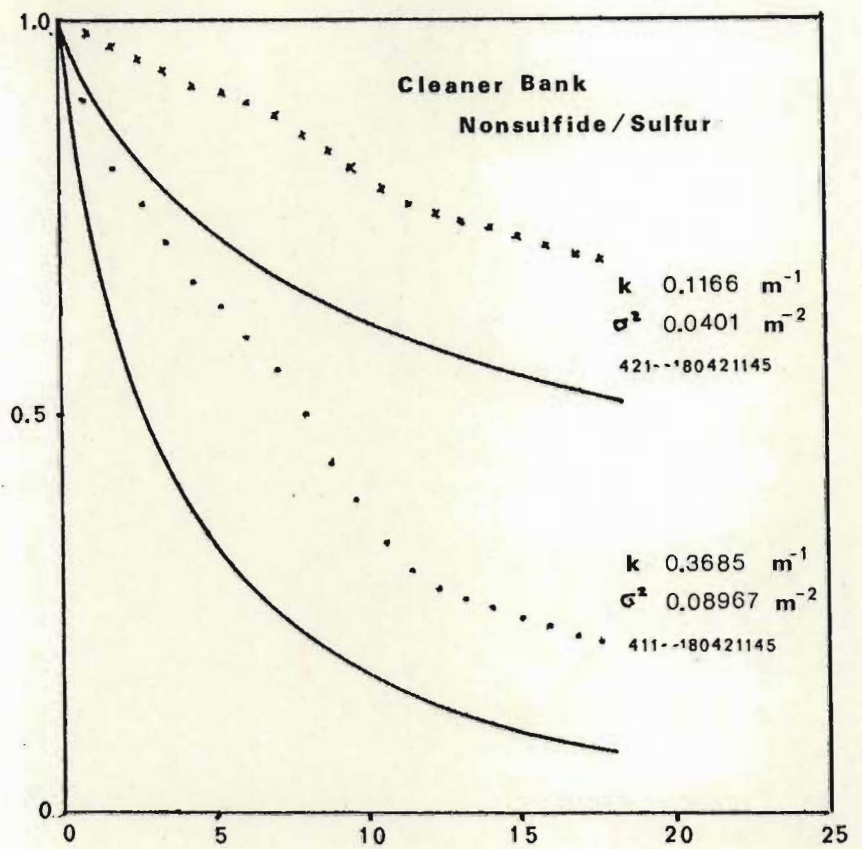
**31. Continuous Tests - Gamma Function Model**

**Fraction in Tailings vs Mean Time - mins**



32. Predicted Continuous Operation - Gamma Function Model

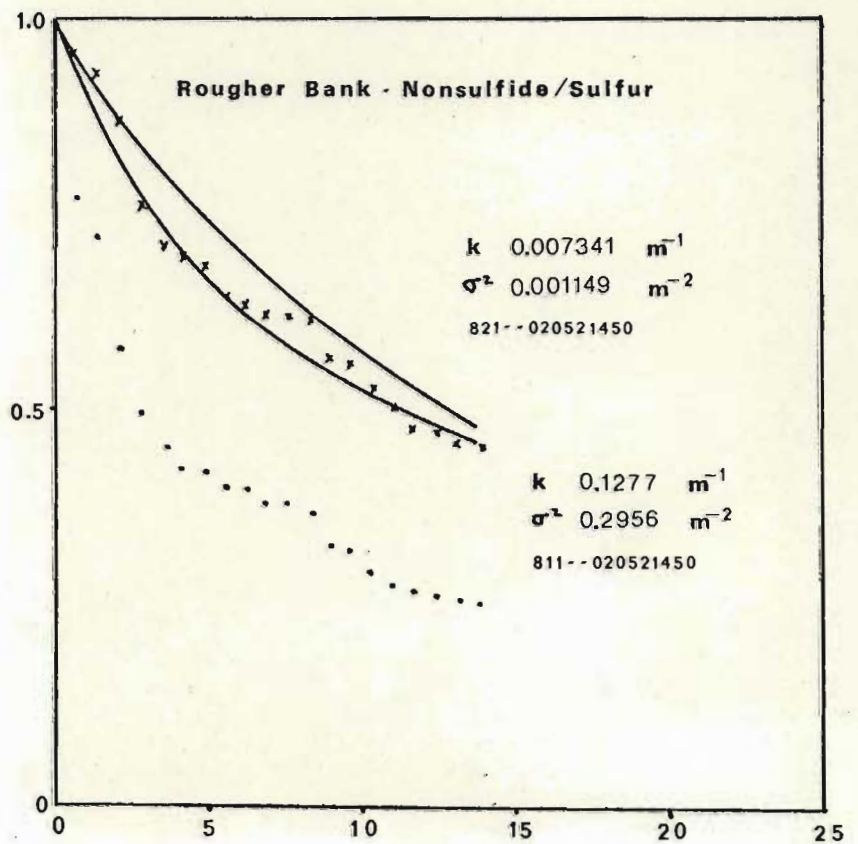
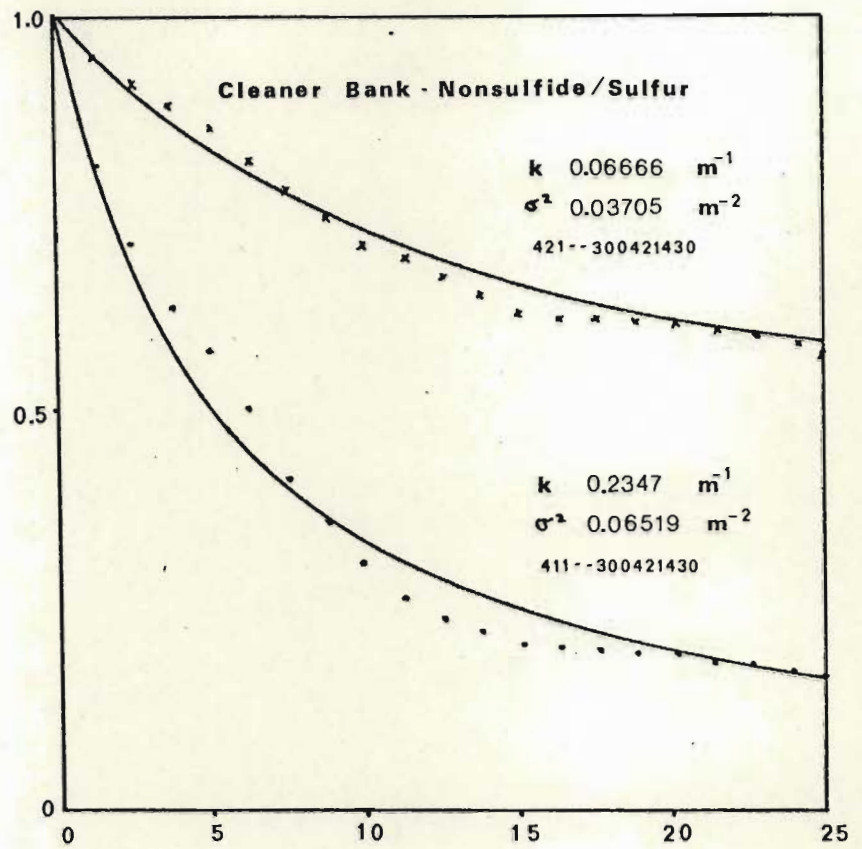
Fraction in Tailings vs Mean Time - mins



33. Predicted Continuous Operation - Gamma Function Model

Fraction in Tailings vs Mean Time - mins





34. Predicted Continuous Operation - Gamma Function Model

Fraction in Tailings vs Mass Time

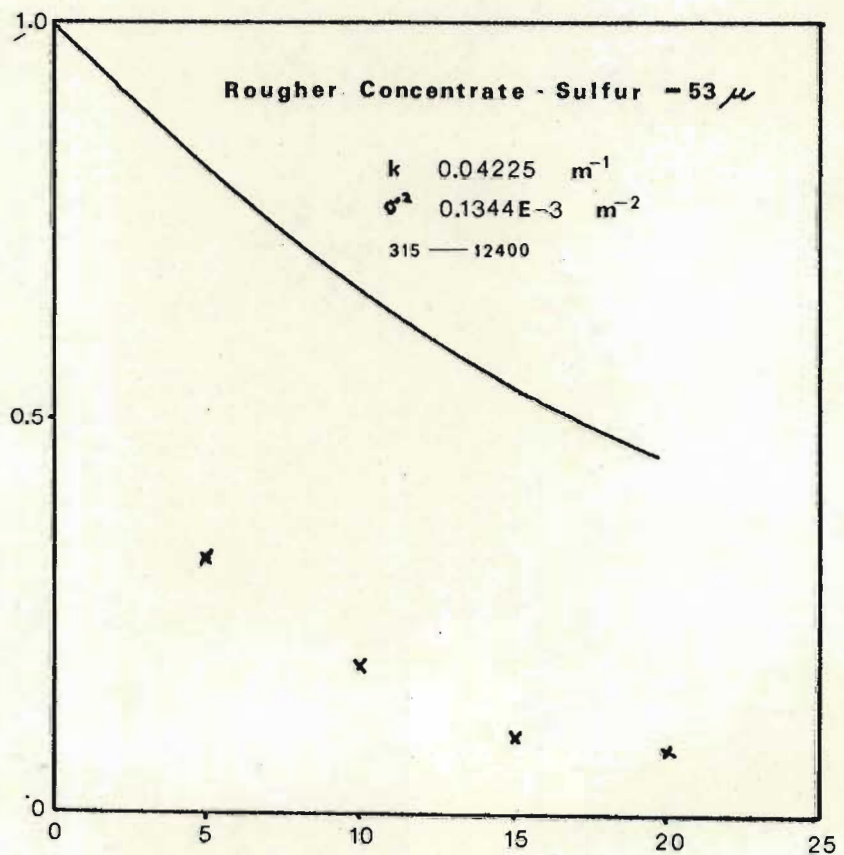
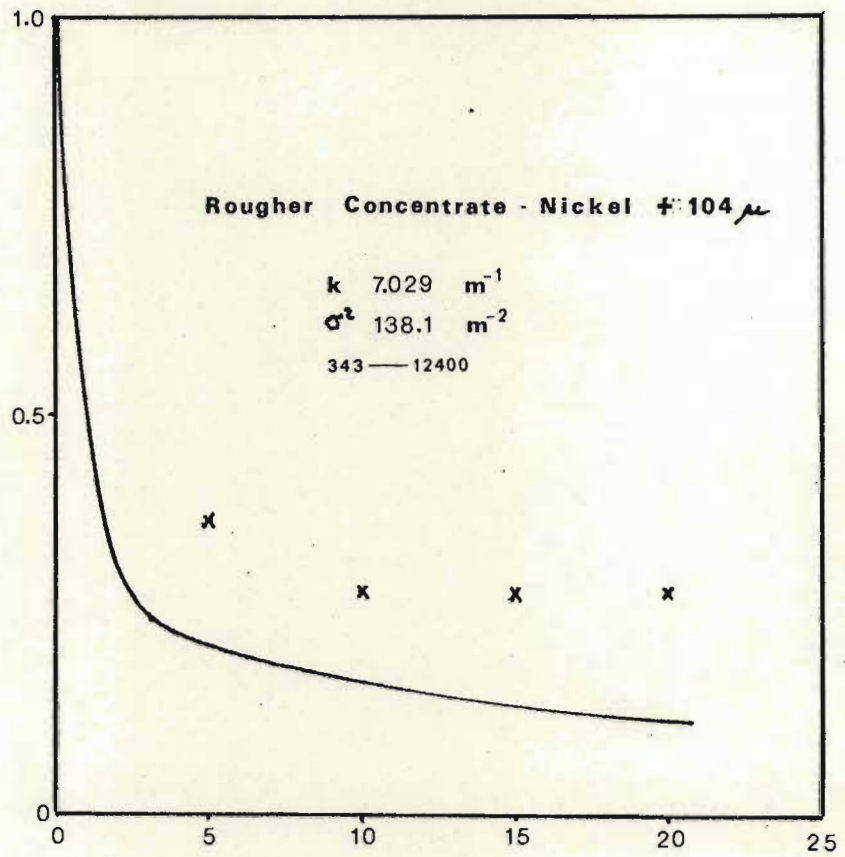
TABLE VI. BATCH TEST PREDICTION OF CONTINUOUS OPERATION -  
GAMMA FUNCTION MODEL

Material and Test	Batch Test Code	Continuous Test Code	Batch Test Parameters			Total Squared Error in Predicted Concentration.
			Mean Rate min <sup>-1</sup>	Variance	Sum of Squares	
Rougher Feed						
Sulphur	811--130411435	811--130421435	0,1083	0,01231	0,01233	0,3933
Non-sulphide	821 "	821 "	0,5821E-2	0,4488E-3	0,2392E-3	0,3941E-3
Copper	831 "	831 "	1,313	3,904	0,9077E-3	0,08356
Nickel	841 "	841 "	0,1588	0,04377	0,8649E-2	0,09997
Rougher Feed						
Sulphur	811--230411200	811--230421200	0,1695	0,04593	0,9985E-2	0,5178
Non-sulphide	821 "	821 "	0,8028E-2	0,1054E-2	0,1982E-3	0,4134E-3
Copper	831 "	831 "	1,533	6,133	0,1055E-2	0,1207
Nickel	841 "	841 "	0,5041	0,7756	0,2933E-2	0,1361
Rougher Feed						
Sulphur	811--020511450	811--020521450	0,1274	0,02944	0,8488E-2	1,010
Non-sulphide	821 "	821 "	0,7143E-2	0,1061E-2	0,6556E-4	0,4977E-3
Cleaner Feed						
Sulphur	411--110411445	411--110421445	0,1256	0,01048	0,2538E-3	0,1765
Non-sulphide	421 "	421 "	0,02833	0,2300E-2	0,1041E-3	0,04736
Copper	431 "	431 "	0,7039	1,080	0,1234E-4	0,03491
Nickel	441 "	441 "	0,1936	0,02913	0,1454E-2	0,1698

TABLE VI. (Continued)

Material and Test	Batch Test Code	Continuous Test Code	Batch Test Parameters			Total Squared Error in Predicted Concentration.
			Mean Rate min <sup>-1</sup>	Variance	Sum of Squares	
Cleaner feed*						
Sulphur	411--180411150	411--180421145	0,3956	0,1146	0,9913E-3	0,9686
Non-sulphide	421 "	421 "	0,1192	0,04261	0,1007E-2	0,6103
Copper	431 "	431 "	1,413	2,927	0,1377E-3	0,6491
Nickel	441 "	441 "	0,6531	0,4570	0,3100E-3	0,5205
Cleaner feed						
Sulphur	411--300411430	411--300421430	0,2462	0,07610	0,2126E-2	0,02963
Non-sulphide	421 "	421 "	0,06643	0,01690	0,7010E-3	0,01478

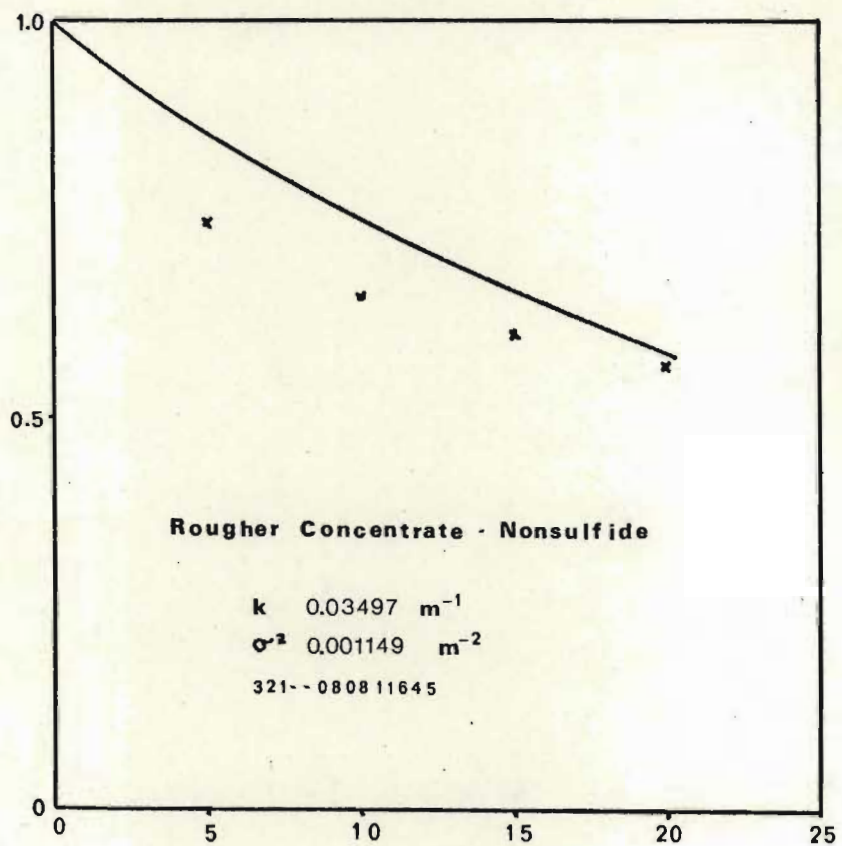
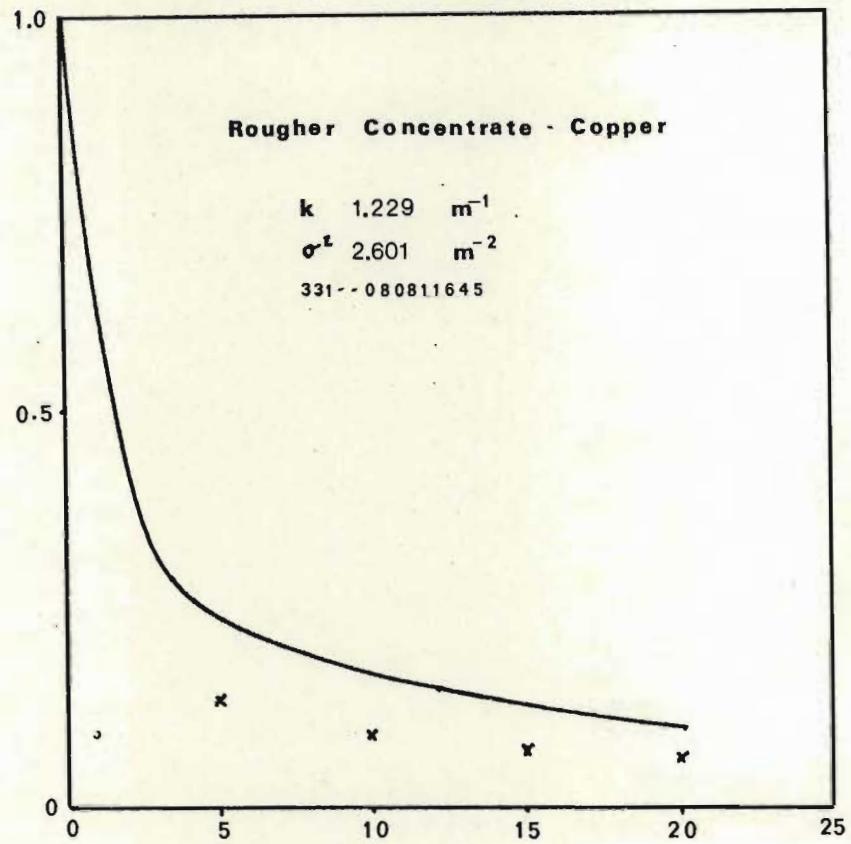




35. Predictions for Refloated Material - Gamma Function Model

Fraction in Tailings vs Time - mins





36. Predictions for Refloated Material - Gamma Function Model

Fraction in Tailings vs Time - mins

TABLE VII. PREDICTIONS FOR REFLOATED MATERIAL -  
GAMMA FUNCTION MODEL

Material	Rougher Feed Test			Rougher Concentrate Test			Rougher Concentrate		Total Squared Error in Concentration Prediction.
	Test Code	Regressed Parameters		Test Code	Regressed Parameters.		Predicted Parameters		
		Mean min <sup>-1</sup>	Variance		Mean min <sup>-1</sup>	Variance	Mean min <sup>-1</sup>	Variance	
Sulphur	811--080311935	0,1857	0,07282	311--080312055	0,2716	0,03898	0,2960	0,09353	0,02198
"	811--010311635	0,4024	0,5815	311--010311800	1,489	4,226	0,7046	0,8401	0,01056
"	811--220211920	0,1179	0,01095	311--220212045	0,7179	0,5837	0,1553	0,01226	0,1306
"	811--200211810	0,2252	0,08312	311--200211945	0,4007	0,07966	0,3208	0,09763	0,04851
"	811--140211805	0,1527	0,03358	311--140211930	0,6737	0,4133	0,2241	0,04017	0,1025
Sulphur (+208 microns)	812-----12400	0,04558	0,05022	312-----12400	3,308	66,11	0,4004	0,3270	0,04047
" (208-104 microns)	813 "	0,1115	0,07693	313 "	0,5243	0,6437	0,3329	0,1740	0,3749E-2
" (104-53 microns)	814 "	0,1657	0,1682	314 "	0,6182	0,5470	0,4486	0,3532	0,02947
" (-53 microns)	815 "	0,03987	0,1290E-3	315 "	0,3205	0,06371	0,04225	0,1344E-3	0,7879
Copper (+ 104 microns)	833 -----12400	0,7056	2,822	333-----12400	7,067	129,6	1,508	5,021	0,02209
" (104-53 microns)	834 "	16,44	1281,	334 "	10,06	184,6	30,95	2037,	0,5805E-3
" (-53 microns)	835 "	0,04902	0,9251E-3	335 "	2,354	5,566	0,06132	0,1068E-2	0,9994
Nickel (+ 104 microns)	843-----12400	0,7895	19,60	343-----12400	4,122	74,87	7,029	138,1	0,09941
" (104-53 microns)	844 "	1,981	36,35	344 "	0,7994	0,7614	6,047	90,51	0,4381E-2
" (-53 microns)	845 "	0,05057	0,1606E-2	345 "	0,7016	0,3919	0,07030	0,1984E-2	0,6890
Sulphur *	811--080811620	0,08825	0,4221E-2	311--080811645	0,4828	0,2723	0,1108	0,4661E-2	0,1459
Non-sulphide *	821 "	0,4090E-2	0,1601E-3	321 "	0,1130	0,05209	0,03497	0,1149E-2	0,02093
Copper *	831 "	0,8467	2,065	331 "	2,600	10,56	1,229	2,601	0,02179
Nickel *	841 "	0,1486	0,04277	341 "	0,7684	0,7361	0,2411	0,05547	0,07503

\* During these tests no dextrin was added to the rougher concentrate.



TABLE VIII. PREDICTIONS FOR REFLOATED MATERIAL -  
GAMMA FUNCTION MODEL WITH NON-FLOATING FRACTION

Material	Rougher Feed Test				Rougher Concentrate			Rougher Con- trate		Total Squared Error in Concentra- tion Pre- diction.
	Test Code	Regressed Parameters			Test Code	Regressed Para- meters		Predicted Parameters		
		Mean min <sup>-1</sup>	Variance	Cn		Mean min <sup>-1</sup>	Variance		Mean min <sup>-1</sup>	
Sulphur	811--080311935	0,1740	0,01316	0,260	311--080312055	0,2716	0,03898	0,1950	0,01325	0,8198E-2
"	811--010811635	0,2468	0,02787	0,333	311--010311800	1,489	4,226	0,2713	0,02772	0,01596
"	811--220211920	0,1154	0,6800E-3	0,156	311--220212045	0,7179	0,5837	0,1179	0,6821E-3	0,1441
"	811--200211810	0,1981	0,01792	0,179	311--200211945	0,4007	0,07966	0,2214	0,01799	0,03283
"	811--140211805	0,1392	0,4527E-2	0,204	311--140211930	0,6737	0,4133	0,1505	0,4555E-2	0,1099
Sulphur (+208 microns)	812-----12400	0,1431	0,1008E-2	0,863	312-----12400	3,308	66,11	0,1455	0,1006E-2	0,2108
" (208-104 microns)	813 "	0,1320	0,8306E-3	0,601	313 "	0,5243	0,6437	0,1343	0,8302E-3	0,05929
" (104-53 microns)	814 "	0,1492	0,3928E-2	0,548	314 "	0,6182	0,5470	0,1578	0,3925E-2	0,03507
" ( -53 microns)	815 "	0,06813	0,2286E-3	0,263	315 "	0,3205	0,06371	0,07014	0,2318E-3	0,3458
Copper (+104 microns)	833-----12400	0,2886	0,02886	0,434	333-----12400	7,067	129,6	0,3048	0,02836	0,02657
" (104-53 microns)	834 "	1,205	0,9863	0,201	334 "	10,06	184,6	1,237	0,9794	0,2372E-2
" ( - 53 microns)	835 "	0,07381	0,1989E-3	0,256	335 "	2,354	5,566	0,07536	0,2007E-3	0,6934
Nickel (+104 microns)	843-----12400	0,2709	0,1392	0,813	343-----12400	4,122	74,87	0,3922	0,1666	0,04192
" (104-53 microns)	844 "	0,2985	0,01571	0,507	344 "	0,7994	0,7614	0,3052	0,1545	0,01221
" ( -53 microns)	845 "	0,08333	0,3306E-3	0,319	345 "	0,7016	0,3919	0,08544	0,3336E-3	0,4091
Sulphur *	811--080811620	0,09586	0,4376E-3	0,148	311--080811645	0,4828	0,2723	0,09801	0,4399E-3	0,1338
Non-sulphide *	821 "	0,05578	0,1850E-3	0,908	321 "	0,1130	0,05209	0,05792	0,1890E-3	0,09616
Copper *	831 "	0,4336	0,1347	0,203	331 "	2,600	10,56	0,4730	0,1342	0,8619E-2
Nickel *	841 "	0,1332	0,3526E-3	0,298	341 "	0,7684	0,7361			**

\* During these tests no dextrin was added to the rougher concentrate.

\*\* The computing facilities available cannot evaluate the necessary integrals for the parameters shown.

### Performance of a Two-Phase Model

The froth phase in flotation is very complex, and in the flotation system it cannot be studied separately from the pulp phase, hence quantitative descriptions of the froth phase are difficult. Its importance in determining recovery and grade for various materials is illustrated in figures 37 and 38; the effect is indeed great. Several points about froth action have been well established by various researchers. Taggart<sup>34</sup> has shown the gradation of values through the froth layer, increased value at the top of the column and relatively poor grades just above the pulp level. The phenomena of upgrading a froth column through increasing its height is a natural function of the particles and system involved. When a froth bubble loaded with particles is disturbed or destroyed, a 'shower' of particles is released and falls back through the froth to the pulp phase. Bubbles may release particles through two primary means; 1) through bubble breakage, and 2) through bubble coalescence (this occurs very rapidly at the bottom of the froth layer). The released particles are carried back into the pulp by water draining from the froth column. The rate of return of particles to the pulp must be a complex function composed not only of the bubble breakage rate, the bubble coalescence rate, the concentration of solids in the froth and the degree of mineral-bubble adherence but also by the froth viscosity, fluid density and by the particle shape. A mathematical description of the sub-processes is too complex for investigation here and the study was limited to determining the importance of the froth phase and the role of frother concentration.

Frothers are chemically composed of heteropolar molecules, the polar portion of the molecule is hydrophilic and the other non-polar portion is hydrophobic. Thus the molecules tend to



concentrate at the air-water interface and lower the surface tension in a pure liquid. Air bubbles in the pulp are reduced in size and a froth is developed. The frothers used at RPM are Aerofroth 71 and cresylic acid, the phenol constituents of cresylic acid may be analyzed for in the various pulp streams by the method outlined in Colorimetric Chemical Analytical Methods.<sup>35</sup> Results of an analysis done on the cleaner bank at RPM are shown in Table X. It is normally assumed that the froth stability results from the concentration of frother, hence the normally voluminous froths in the cleaner sections of networks. This assumption is not supported by the cresylic acid analyses, however, as the majority of the frother remains in the tailings stream. Froth stability is also dependent upon the solids portion of the pulp, particularly the fine material. Gaudin<sup>50</sup> states that even in aqueous solutions of maximum frothing ability, the froth produced is very short lived. In a mineral suspension, however, the froth produced may last for several hours. It is not probable that the alcohol frother, AF71, acts in a different manner from the cresylic acid. Hence the main reason for stable and voluminous froths in the cleaner section would appear to be the nature of the solid material rather than the frother concentration. In example, in a typical plant situation at RPM, the rougher feed would contain approximately 50 grams/min (30 g/ton) of cresylic acid, the cleaner feed stream 6 grams/min (12,86 g/ton) and the final concentrate less than 0,2 grams/min. (4 g/ton). It is obvious from this data then that low froth production at the end of a rougher bank is not a result of lack of frother but more from a lack of floatable material which gives the froth its stability.

Several indications of the importance of the froth phase are found in the preceding sections. As implied by Arbiter

and Harris, the apparent rate of flotation reflects the froth removal rate in any instance where material appears in the froth phase more rapidly than the froth is removed. In general, flotation in the first few cells and in the cleaners is governed not by the kinetics of flotation but rather by the 'pull'; on the other hand, near the end of a bank of cells the process is kinetically determined.

An examination of the single-phase kinetic flotation models of previous chapters has proved them unsuitable mathematically for description of true flotation where the kinetic parameters are such that the froth removal rate has a marked effect on the flotation of a species, as in the cleaner section at RPM. This problem may be handled in two basic ways; one, the single phase kinetic parameters may be extended as not only functions of the species but also as functions of the froth removal rate, and two, a second phase (the froth phase) may be included in the kinetic formulation of the system and the resulting mathematical description.

Arbiter and Harris,<sup>36</sup> and Harris and Rimmer<sup>37</sup> have chosen the second of these methods. The two-phase model proposed by Arbiter and Harris in 1966, assumes both the pulp and the froth phases to be 'well mixed' and consistent with first-order kinetics. Thus for any chosen species a single rate constant will describe the transfer of the species from the pulp to the froth and a second first-order rate constant will describe the transfer of the species in the reverse direction. It has already been shown that for any mineralogical or metal species, a fraction of the species is non-floatable; addition of this fraction is expected to improve the accuracy of the models fit.

A large number of batch flotation tests, in which the removal rate of concentrate as well as the froth depth were recorded, were performed. A summary of the available data



which includes the information for a bank of continuously operating cells is presented in Tables IX and X. The flow-rate of froth was measured as the wet weight of concentrate per unit time and converted to volume flow per unit time. The volume of froth was calculated from cell dimensions and froth depth thus it includes the volume of air entrained in the froth.

#### Batch Test Results

Due to the close relationship between the froth phase and the pulp phase in flotation, it would be expected that the parameters involved would be very interdependent. Figure 39 shows the best fit of the two-phase model, a smooth curve is not fitted since the  $Q_f/V_f$  value varies from point to point, and figure 40 shows the sum of squares surfaces respectively. Note the very poorly defined minimums where the sum of squares surface changes very little for changes in 'a' and 'b'.

Since the model incorporates the froth phase, parameters regressed from one test were used in predicting other tests. Figure 41 illustrates this. In figure 41.1 parameters were obtained from a test with high  $Q_f/V_f$  values and used to predict a test with lower  $Q_f/V_f$  values; in figure 41.2 the opposite was done. These figures indicated that the model attaches too great a significance to the  $Q_f/V_f$  values.

The model's predictive capacity is improved by the addition of a non-floating fraction of the species. Figure 42 is an example of this. Parameter interdependence is still a serious problem however, as indicated by the sum of squares surface in figure 43.1. Predictions for different  $Q_f/V_f$

values still show the model's over-emphasis of the froth residence times as in figure 43.2, where 42.1 is predicted from 42.2.

### Conclusions

The two-phase model as described by Harris and Rimmer is unsatisfactory in handling batch data. The addition of a non-floating fraction of the species improves the model but parameter significances are very poor. The value of  $Q_f/V_f$  is given more importance in the model than what is suggested by the data and thus the model overreacts to changes in  $Q_f/V_f$ .



TABLE IX. FROTH REMOVAL DATA - BATCH TESTS

Description and Test Code	Flotation Time (min)							Variable
	5	10	15	20	30	40	50	
Rougher Feed	0,0787	0,0772	0,139	0,314	0,478	0,480		Q/V min <sup>-1</sup>
811--110212040	0,652	0,575	0,435	0,393	0,339	0,303		S
821 "	0,988	0,980	0,971	0,965	0,956	0,948		NS
Rougher Feed	0,104	0,0801	0,0697	0,0714	0,0637	0,0645		Q/V
811--130211915	0,762	0,632	0,576	0,538	0,477	0,431		S
821 "	0,993	0,988	0,982	0,978	0,970	0,963		NS
Rougher Feed	0,279	0,261	0,320	0,233	0,334	0,280		Q/V
811--140211805	0,571	0,494	0,354	0,304	0,236	0,204		S
821 "	0,970	0,950	0,927	0,914	0,895	0,882		NS
Rougher Feed	0,752	0,626	1,40	1,18	2,20	3,40		Q/V
811--200211810	0,484	0,464	0,311	0,254	0,202	0,179		S
821 "	0,955	0,944	0,914	0,894	0,873	0,859		NS
Rougher Feed	1,495	1,065	1,079	0,730	0,634	0,531		Q/V
811--220211920	0,606	0,470	0,310	0,245	0,183	0,156		S
821 "	0,968	0,942	0,919	0,905	0,884	0,873		NS
Rougher Feed	0,0733	0,0593	0,0607	0,0624	0,0597	0,0698		Q/V
811--260212030	0,732	0,657	0,551	0,504	0,436	0,407		S
821 "	0,992	0,986	0,981	0,976	0,969	0,964		NS
Rougher Feed	0,667	0,461	2,220	1,52	0,645	0,546		Q/V
811--010311625	0,550	0,512	0,401	0,351	0,287	0,260		S
821 "	0,967	0,954	0,932	0,921	0,905	0,893		NS
Rougher Feed	1,92	1,45	1,42	1,00	0,746	0,618		Q/V
811--080311935	0,574	0,512	0,401	0,351	0,287	0,260		S
821 "	0,966	0,952	0,934	0,923	0,907	0,898		NS
Rougher Feed	0,336	0,354	0,700	0,837	0,547	0,341		Q/V
811--130411435	0,633	0,565	0,336	0,281	0,244	0,230		S
821 "	0,980	0,967	0,938	0,924	0,911	0,905		NS
831 "	0,288	0,245	0,174	0,153	0,134	0,124		CU
841 "	0,626	0,495	0,326	0,305	0,290	0,284		NI
Rougher Feed	0,727	0,592	0,545	0,359	0,237	0,185		Q/V
811--230411200	0,560	0,518	0,328	0,273	0,248	0,237		S



TABLE IX. (Continued)

Description and Test Code	Flotation Time (min)							Variable
	5	10	15	20	30	40	50	
821--230411200	0,971	0,959	0,931	0,916	0,906	0,899		NS
831 "	0,301	0,268	0,200	0,174	0,157	0,148		CU
841 "	0,489	0,429	0,321	0,298	0,286	0,280		NI
Rougher Conc.	0,0162	0,0185	0,0151	0,0175	0,0217	0,0245		Q/V
311--110212145	0,775	0,576	0,419	0,319	0,196	0,111		S
321 "	0,947	0,893	0,831	0,776	0,686	0,617		NS
Rougher Conc.	0,0235	0,0230	0,0240	0,0231	0,0231	0,0223		Q/V
311--130212035	0,612	0,404	0,294	0,215	0,177	0,100		S
321 "	0,960	0,924	0,896	0,868	0,839	0,799		NS
Rougher Conc.	0,534	0,443	1,010	1,13				Q/V
311--140211930	0,220	0,104	0,079	0,069				S
321 "	0,805	0,703	0,650	0,617				NS
Rougher Conc.	0,598	0,375	0,439	0,390				Q/V
311--200211945	0,251	0,105	0,061	0,046				S
321 "	0,751	0,635	0,557	0,520				NS
Rougher Conc.	0,516	0,429	0,687	0,905				Q/V
311--220212045	0,242	0,126	0,103	0,091				S
321 "	0,752	0,647	0,598	0,566				NS
Rougher Conc.	0,0540	0,0615	0,0618	0,0584	0,0572	0,0659		Q/V
311--260212155	0,756	0,580	0,496	0,394	0,237	0,162		S
321 "	0,962	0,929	0,899	0,872	0,811	0,770		NS
Rougher Conc.	0,954	0,869	1,18	1,025				Q/V
311--010311800	0,247	0,161	0,138	0,126				S
321 "	0,839	0,773	0,741	0,718				NS
Rougher Conc.	1,00	1,13	3,29	0,961				Q/V
311--080312055	0,369	0,166	0,110	0,095				S
321 "	0,811	0,699	0,631	0,600				NS
Cleaner Feed	0,0394	0,0528	0,0693	0,0813	0,0889	0,134	0,252	Q/V
411--110411445	0,590	0,407	0,305	0,232	0,144	0,111	0,095	S
421 "	0,894	0,815	0,754	0,711	0,645	0,606	0,571	NS
431 "	0,270	0,168	0,127	0,106	0,078	0,064	0,053	CU
441 "	0,516	0,294	0,205	0,162	0,110	0,088	0,074	NI

TABLE IX (Continued)

Description and Test Code	Flotation Time (min)							Variable
	,5	10	15	20	30	40	50	
Cleaner Feed	0,270	0,512	0,630	0,532	0,407			Q/V
411--180411150	0,312	0,136	0,091	0,074	0,058			S
421 "	0,729	0,588	0,526	0,492	0,456			NS
431 "	0,195	0,115	0,090	0,078	0,066			CU
441 "	0,254	0,132	0,096	0,080	0,064			NI

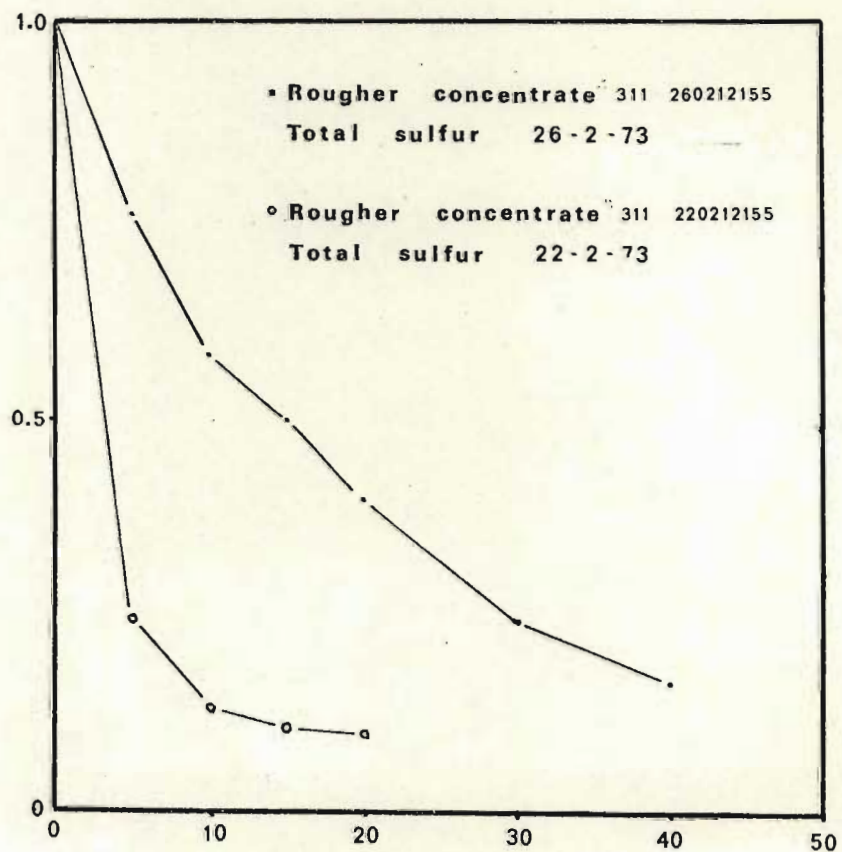
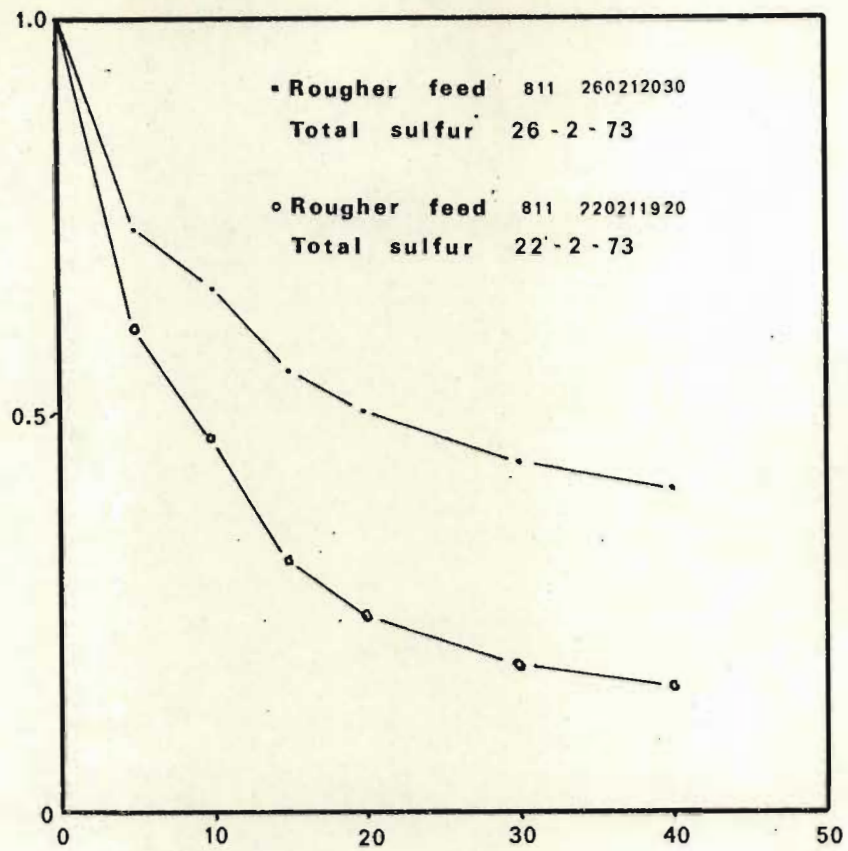
\* Q/V is the flowrate of froth divided by the froth volume.  
S, NS, CU, NI symbolize the fraction of sulphur, non-sulphide,  
copper and nickel in the tailings.



TABLE X. FROTH REMOVAL DATA - CONTINUOUS TESTS

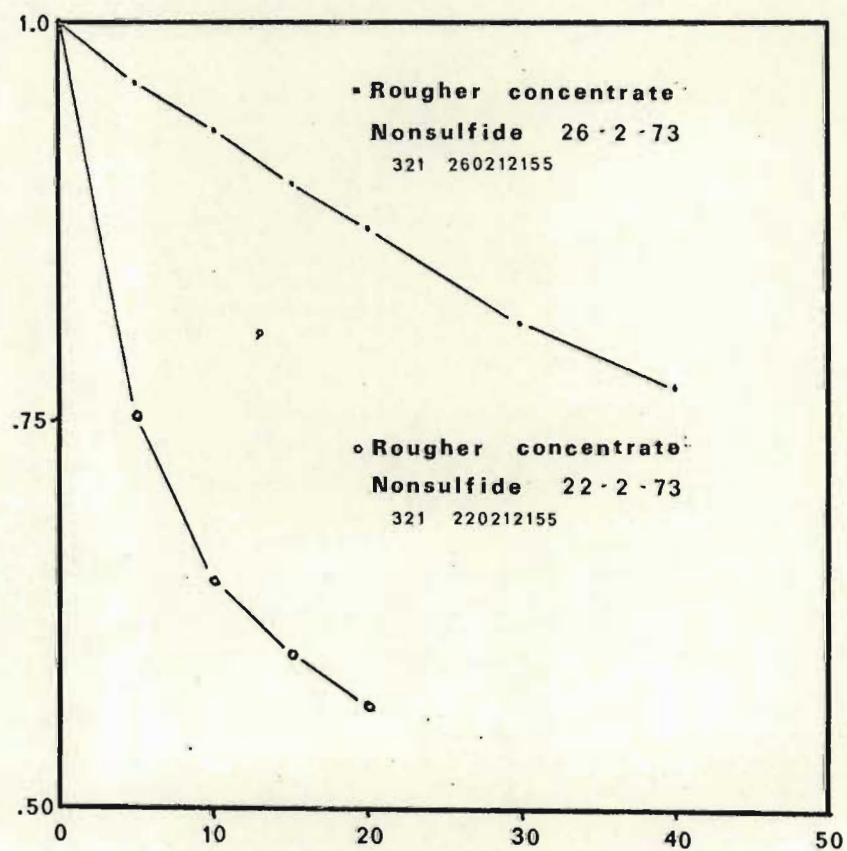
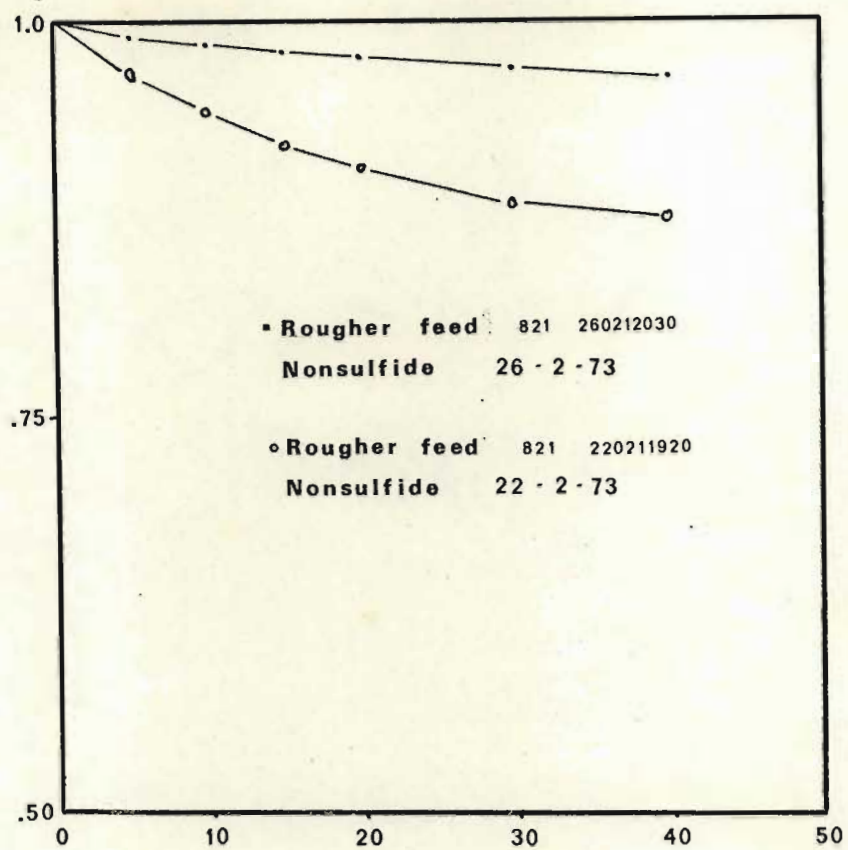
		Cleaner Bank 411--110421445							Cleaner Bank 411--300421430						
		Mean Time (min)	Sulphur Fraction	Nonsulphide Fraction	Copper Fraction	Nickel Fraction	Froth Flow Kg/min	Froth % Solids	Cresylic Acid Flowrate g/min	Mean Time	Sulphur Fraction	Nonsulphide Fraction	Froth Flow Kg/min	Froth % Solids	Froth Depth (cms)
1		0,0	1,000	1,000	1,000	1,000	-		5,92	0,0	1,000	1,000	-	-	-
	1	0,97	0,862	0,978	0,681	0,869	10,16	27,9	0,028	1,25	0,829	0,957	15,3	39,5	36
	2	1,95	0,808	0,970	0,578	0,814	3,95	28,9	0,012	2,51	0,719	0,923	12,2	38,0	"
	3	2,93	0,752	0,959	0,479	0,742	5,52	33,0	0,006	3,76	0,637	0,889	14,8	29,9	"
	4	3,90	0,642	0,937	0,349	0,590	11,90	31,0	0,005	5,02	0,582	0,862	10,2	33,7	"
	5	4,88	0,524	0,904	0,252	0,434	19,89	25,3	0,004	6,29	0,507	0,823	21,6	23,2	34
	6	5,87	0,474	0,884	0,213	0,371	13,85	22,3	0,009	7,54	0,420	0,780	21,9	25,3	"
	7	6,84	0,438	0,868	0,189	0,326	9,35	25,6	0,003	8,80	0,366	0,747	13,7	30,1	"
	8	7,82	0,392	0,846	0,168	0,281	14,79	22,0	0,010	10,06	0,316	0,713	19,8	21,6	"
	9	8,80	0,357	0,822	0,150	0,246	14,92	22,5	-	11,31	0,269	0,695	12,0	19,4	22
	10	9,78	0,329	0,801	0,138	0,222	14,75	20,5	-	12,58	0,242	0,671	13,1	22,5	"
	11	10,76	0,307	0,783	0,128	0,204	13,13	20,4	-	13,83	0,225	0,653	8,6	26,1	"
	12	11,74	0,271	0,756	0,116	0,181	19,51	19,6	-	15,09	0,204	0,626	14,1	22,9	"
	13	12,72	0,239	0,733	0,104	0,163	21,12	16,0	-	16,35	0,203	0,623	1,1	25,8	26
	14	13,69	0,217	0,712	0,095	0,149	22,57	12,9	-	17,60	0,201	0,621	0,9	34,0	"
	15	14,67	0,199	0,693	0,087	0,137	19,55	14,0	-	18,87	0,200	0,619	0,9	19,1	"
	16	15,66	0,173	0,668	0,078	0,123	25,42	13,9	-	20,13	0,195	0,610	3,2	34,8	"
	17	16,63	0,169	0,663	0,076	0,120	4,08	18,1	-	21,38	0,189	0,600	4,9	23,7	24
	18	17,61	0,168	0,661	0,075	0,118	1,53	21,6	-	22,64	0,184	0,593	3,5	23,9	24
	19	18,59	0,167	0,659	0,074	0,117	1,70	15,3	-	23,90	0,176	0,583	5,1	23,6	"
	20	19,57	0,162	0,650	0,070	0,112	6,37	19,6	-	25,16	0,164	0,567	8,7	22,5	"





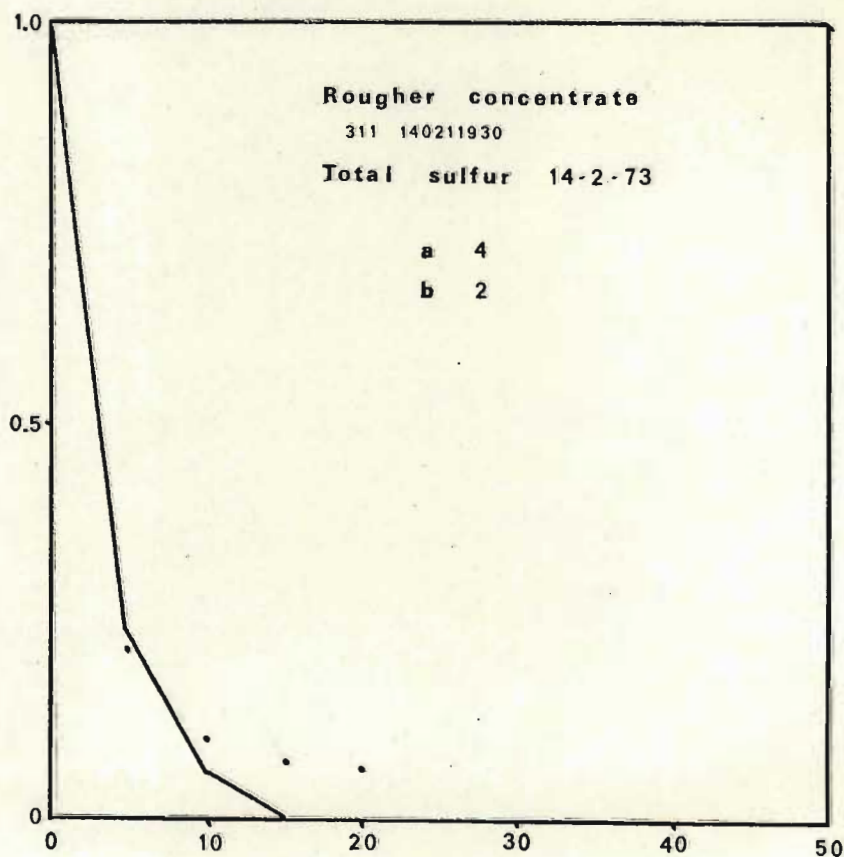
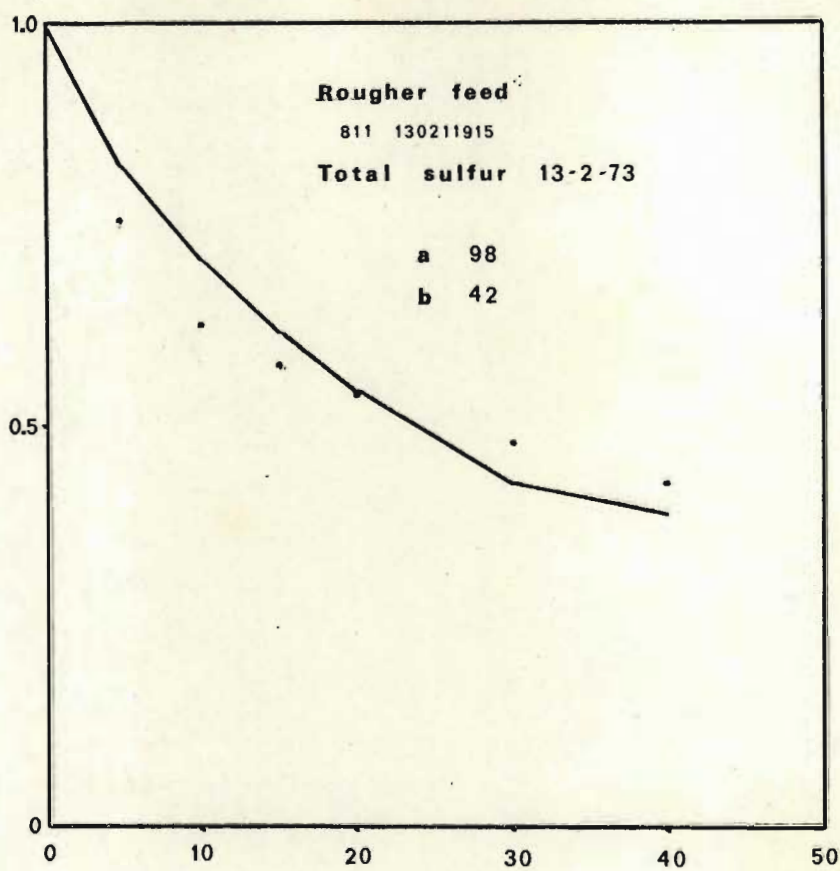
### 37. Importance of the Froth Phase

Fraction in Tailings vs. Time - mins



38. Importance of the Froth Phase

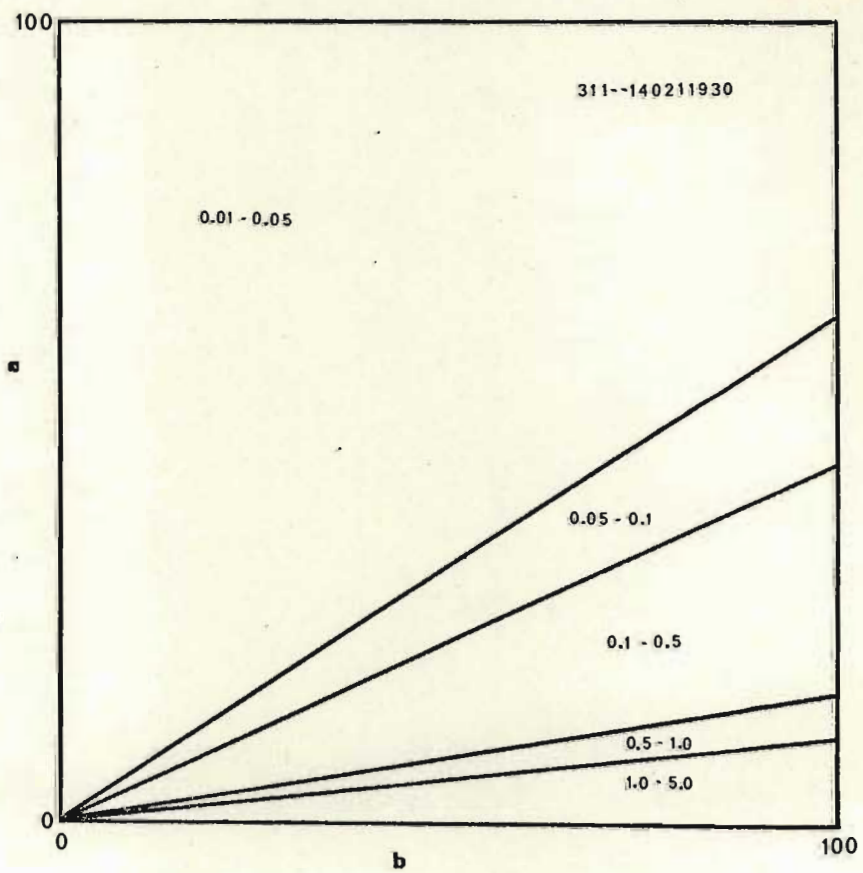
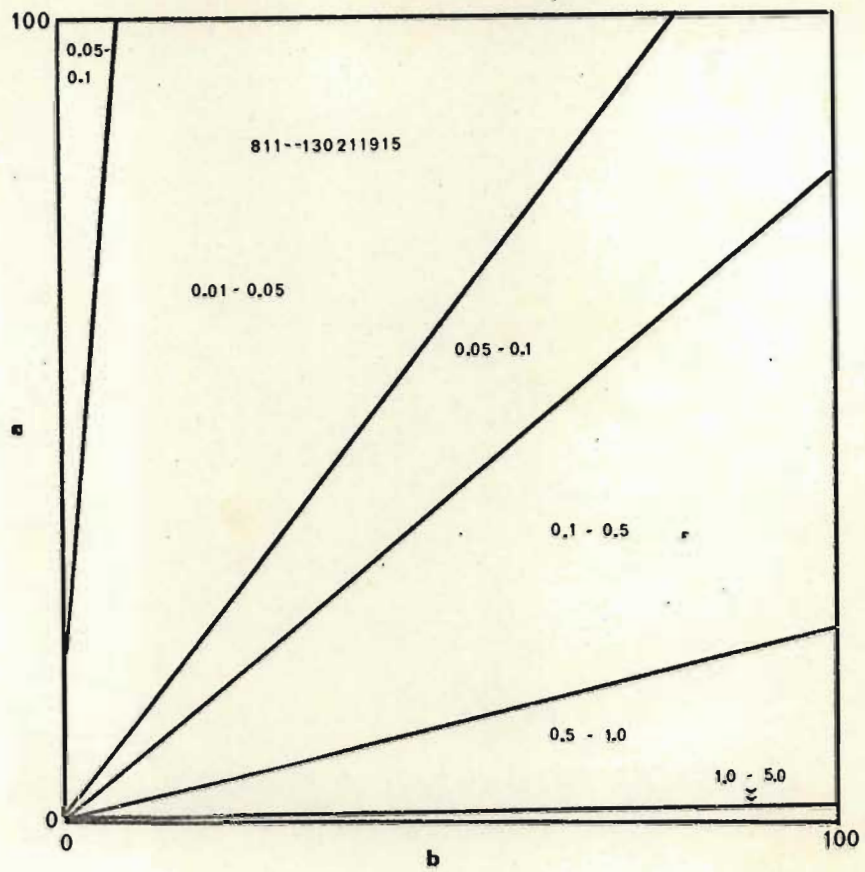
Fraction in Tailings vs. Time - mins

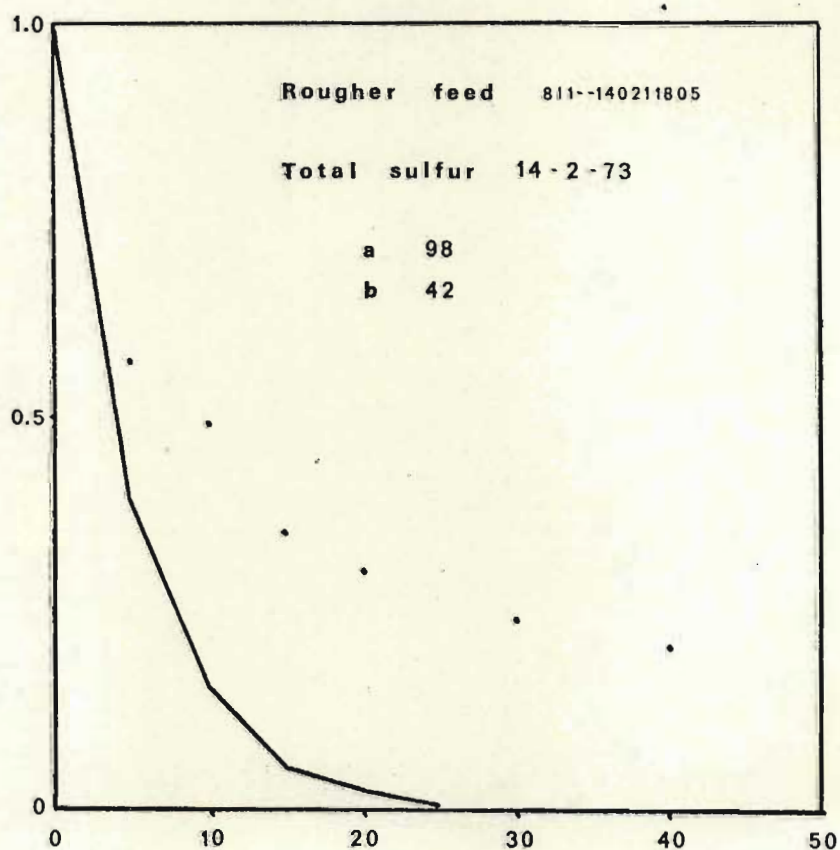
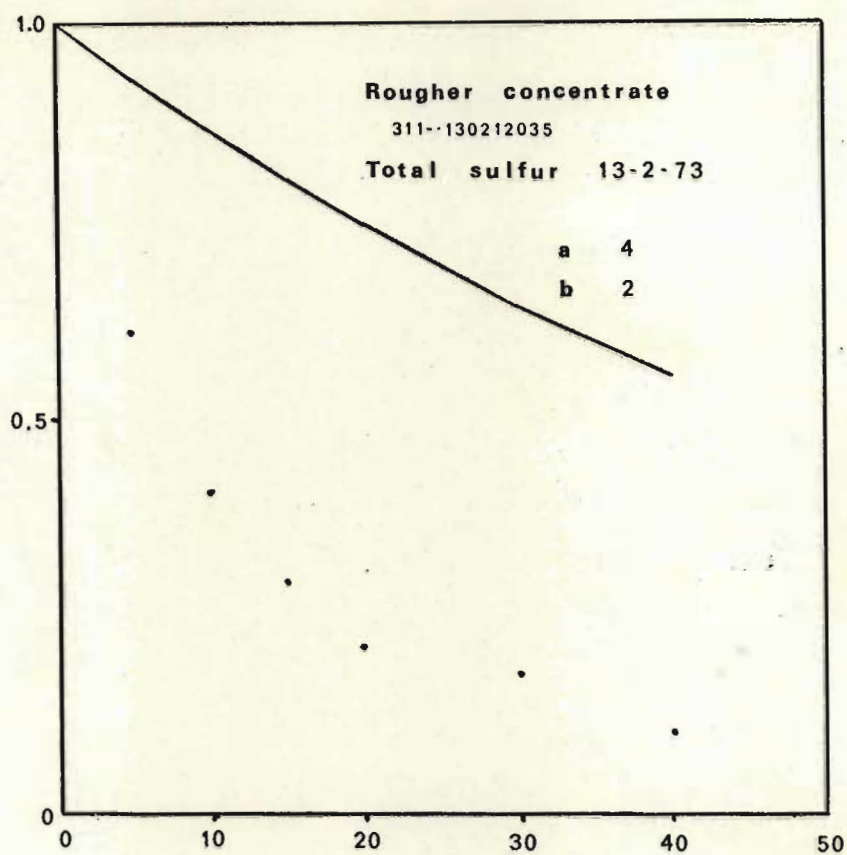


39. Batch Tests - Two Phase Model

Fraction in Tailings vs. Time - mins

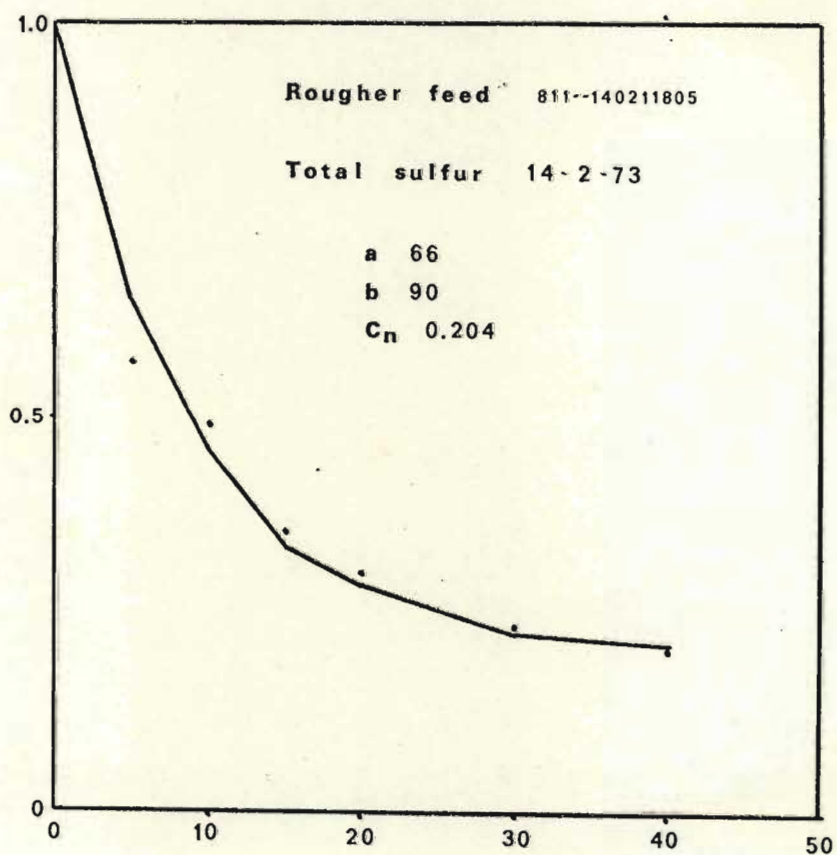
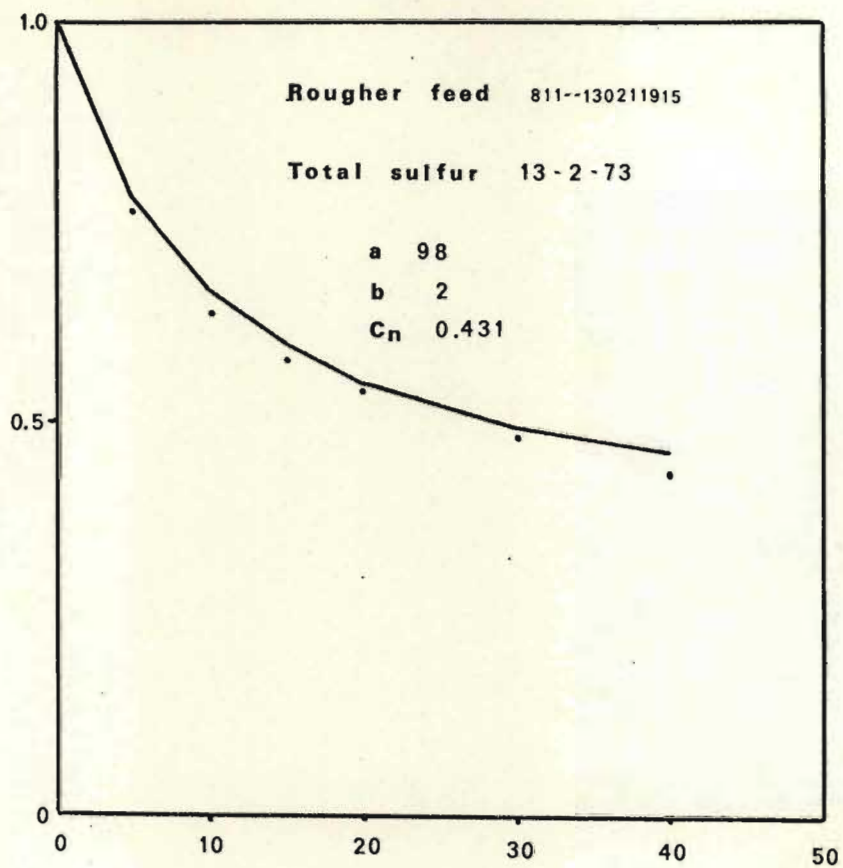






41. Two Phase Model Predictions

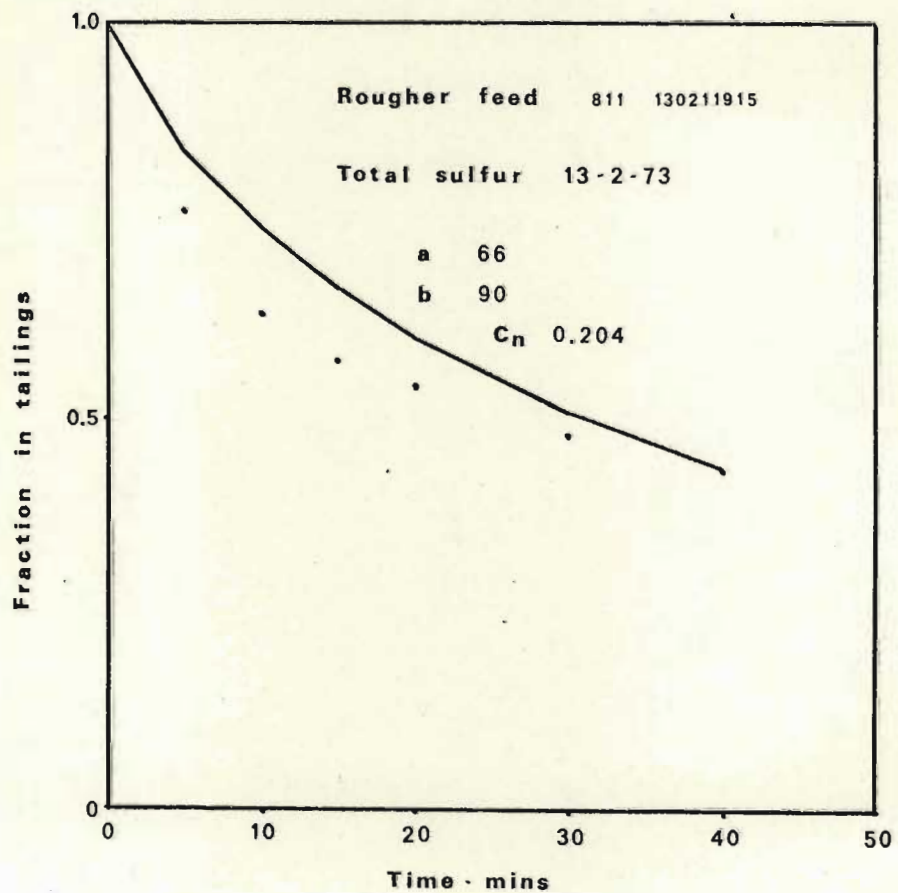
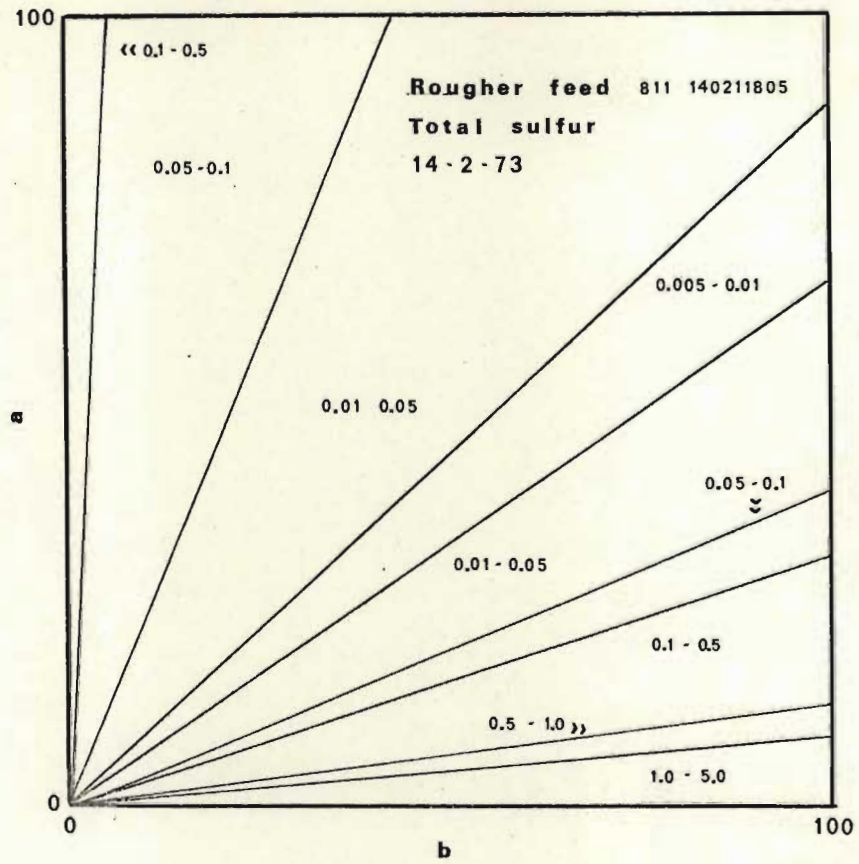
Fraction in Tailings vs. Time - mins



42. Batch Tests - Two Phase Model

Fraction in Tailings vs. Time - mins





43. Two Phase Model - Sum of Squares Surface  
and Batch Prediction

### CONCLUSIONS ABOUT THE FOUR FLOTATION MODELS

Up to this point a considerable amount of data regarding the performance of four chemical analogy models in batch operation, continuous operation, batch prediction of continuous operation and prediction of the refloatation of materials has been presented. Several generalisations may be made about the applicability of these models before proceeding toward the difficult stage of circuit design analysis.

Clearly the single rate first-order model is unsuited for any application, whether it be the simple description of batch flotation of the more complicated predictive requirements. A single rate constant cannot adequately describe the flotation behaviour of the very individual particles in a mineralogical or metal species. The factors of shape and surface condition vary for any mineralogical species to a degree which leads to gross inadequacies for the single rate model.

The two-phase first order model is also severely limited. Although the model has the attraction of including the froth phase mathematically, the parameter interaction and general poor parameter definition as well as an over-reaction to changes in the froth flowrate make the model presently unsuitable. The description of species flotation by a single rate parameter (transfer from pulp to froth) has been proved lacking in the case of the single rate model; the addition of a non-floating fraction improves the model's batch description but does not alleviate the other shortcomings. It is likely that any flotation model which attempts to directly incorporate the actions of the froth phase will suffer from the problem of parameter interdependence. In flotation, the pulp and froth phase are experimentally difficult to separate and their examination together will always lead to poor parameter estima-



tions. The overall complexity of the froth phase and its close dependence upon the pulp phase make froth modelling a difficult problem to attack realistically. However the froth phase, particularly the froth removal rate, is of such importance that any realistic flotation model cannot fail to incorporate froth behaviour in its application.

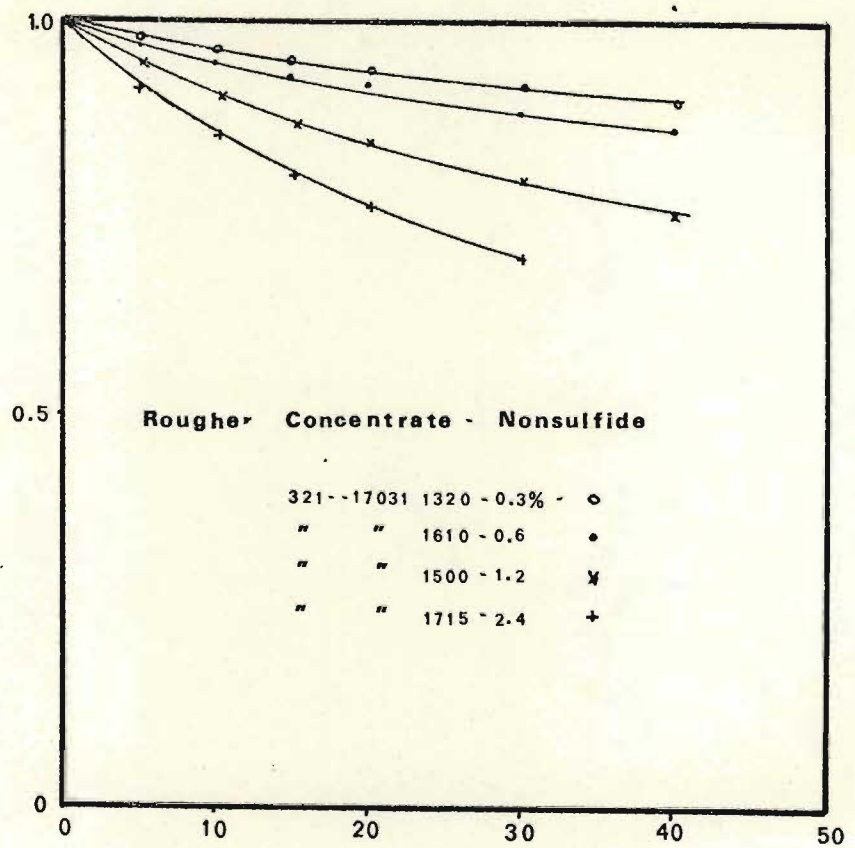
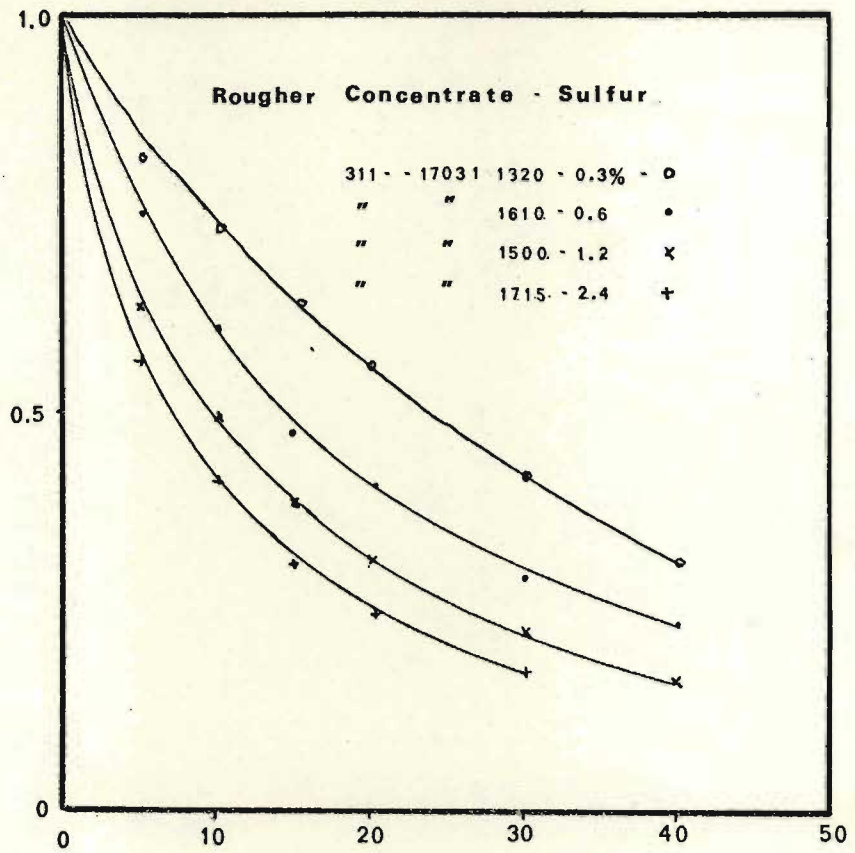
The other two single phase models, the rate plus non-floating model and the gamma function model, may be made to react to frothing conditions if their kinetic parameters prove to be reasonable functions of the froth removal rate. The data from figures 44 and 45 is used to present the kinetic parameters  $\bar{k}$  (mean) and  $\sigma^2$  (variance) as functions of the froth removal rate in figures 46 and 47. The rate distribution parameters appear to change smoothly with the change in froth removal rate, except for nickel material. Movement of the parameters for the rate plus non-floating model is not so consistent, partially because the non-floating fraction is not always distinct. Theoretically, however, the non-floatable fraction in this case should not alter with the froth removal rate. This abnormality clearly places the rate plus non-floating model at a disadvantage as can be seen by the figures in Table XI.

In addition to the complications of the froth phase, two other problems were evident in the past data analysis. Firstly, the regressed parameters for any model are equipment-specific. This is illustrated by the predictions of continuous operation with the batch test parameters; while the sulphide predictions are poor, predictions of copper and nickel are within the limits of froth behaviour. Either model, the rate plus non-floating or the gamma function model, is adequate but the gamma function model appears to predict the bulk of the continuous operations more accurately.



Secondly, the assumed time invariancy of a particle's floatability is questionable. Certainly both models fall short of accurate prediction of the rougher concentrate flotation from the kinetic parameters of the rougher feed; the species normally floats faster than predicted. The formation of the rate plus non-floating model leads to large discrepancies where once floated material exhibits a non-floating fraction. The addition of a non-floating fraction and the description of the floatable material by the gamma function rate constant distribution is more plausibly the true situation, and does provide slightly superior results. Still, the majority of industrial works add reagents in several stages for the express purpose of altering the particles' floatability, it can only be hoped that such an effect is minimal within the broader scope of the modelling objectives.

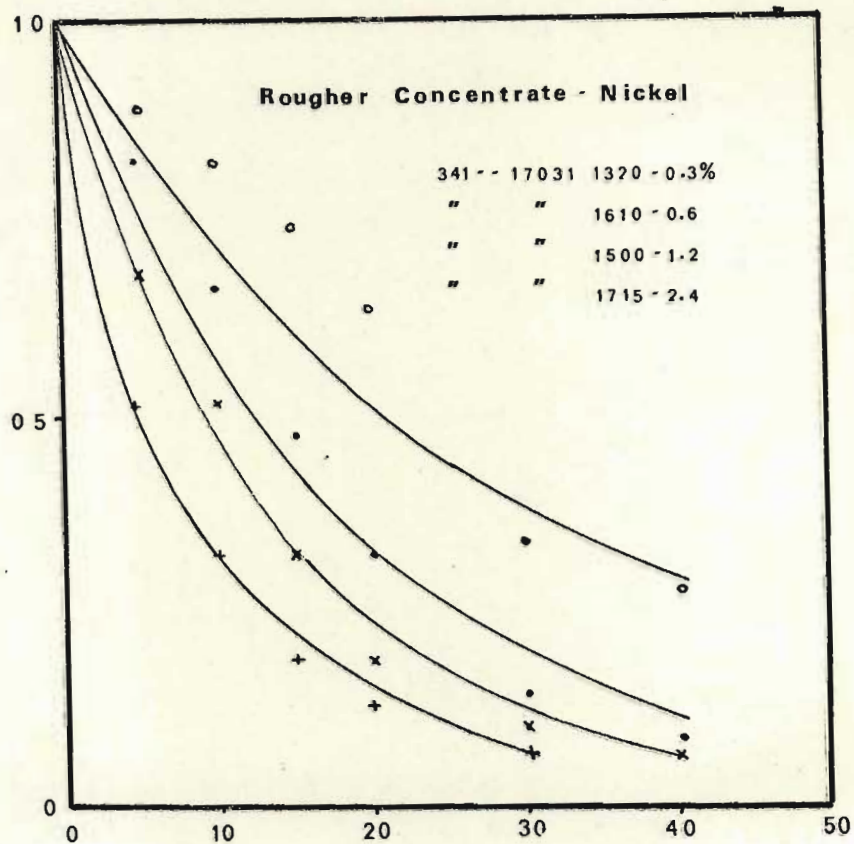
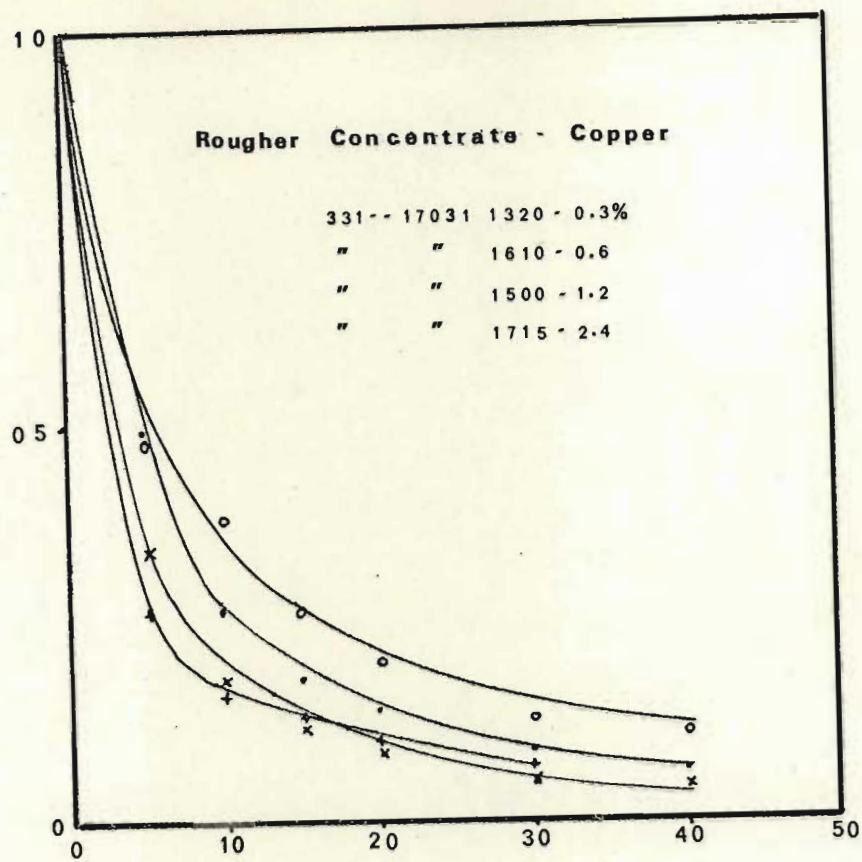
The gamma function model clearly possesses several advantages and provides the most plausible framework for the development of a production-design tool. It is inherently more descriptive of the individual particles within a mineralogical species and has shown itself to be slightly more accurate in the description of batch data, and continuous data as well as being more accurate in a predictive capacity. Refer to Tables XII, XIII and XIV. As the gamma function model is mathematically capable of describing the operation of various flotation networks and has shown promise, its applicability to the analysis of an industrial flotation circuit will be examined in the following chapter.



#### 44. Batch Tests - Froth Removal Rate

Gamma Function Model

Fraction in Tailings vs Time - mins

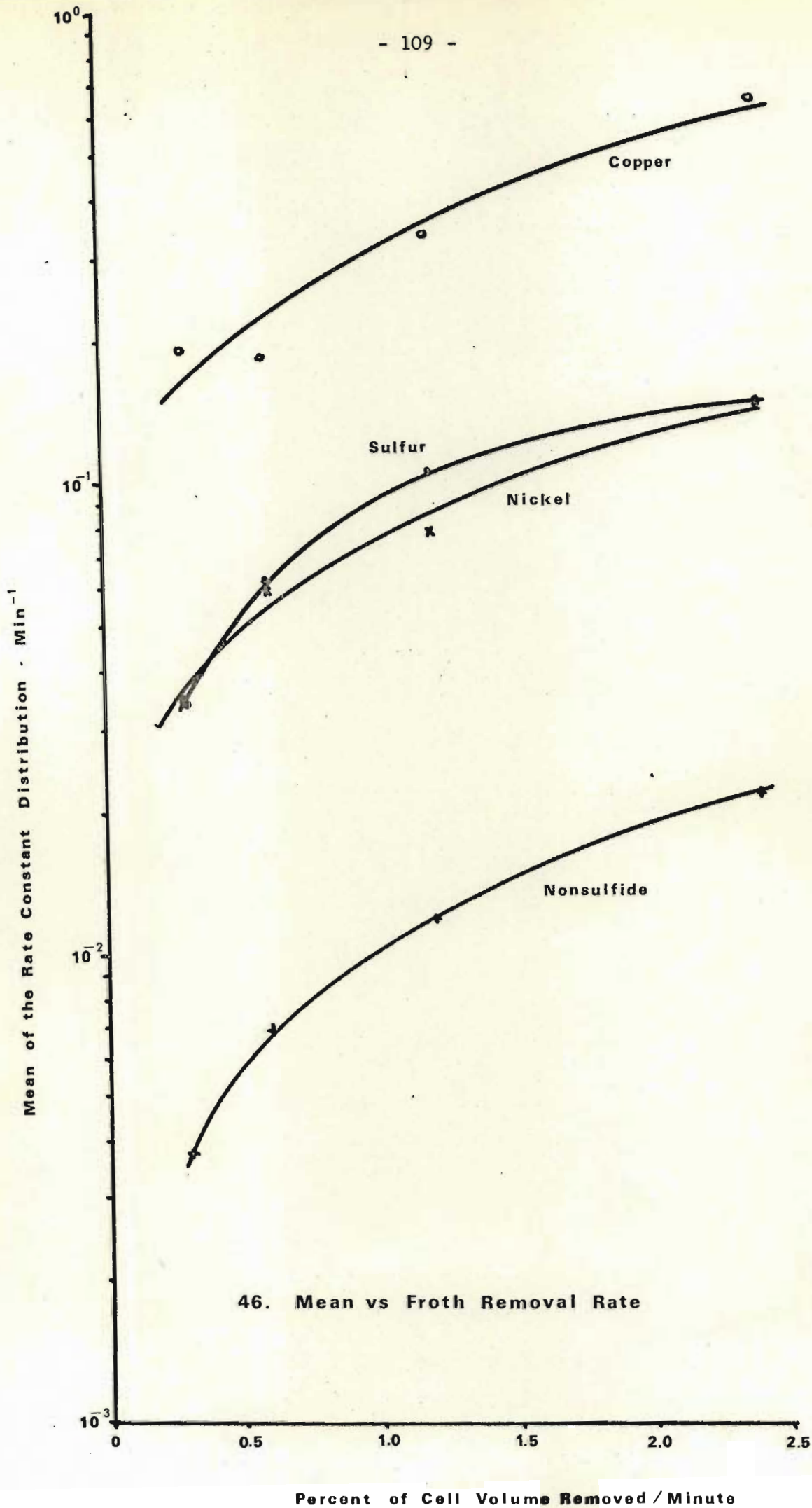


#### 45. Batch Tests - Froth Removal Rate

**Gamma Function Model**

**Fraction in Tailings vs Time - mins**





46. Mean vs Froth Removal Rate

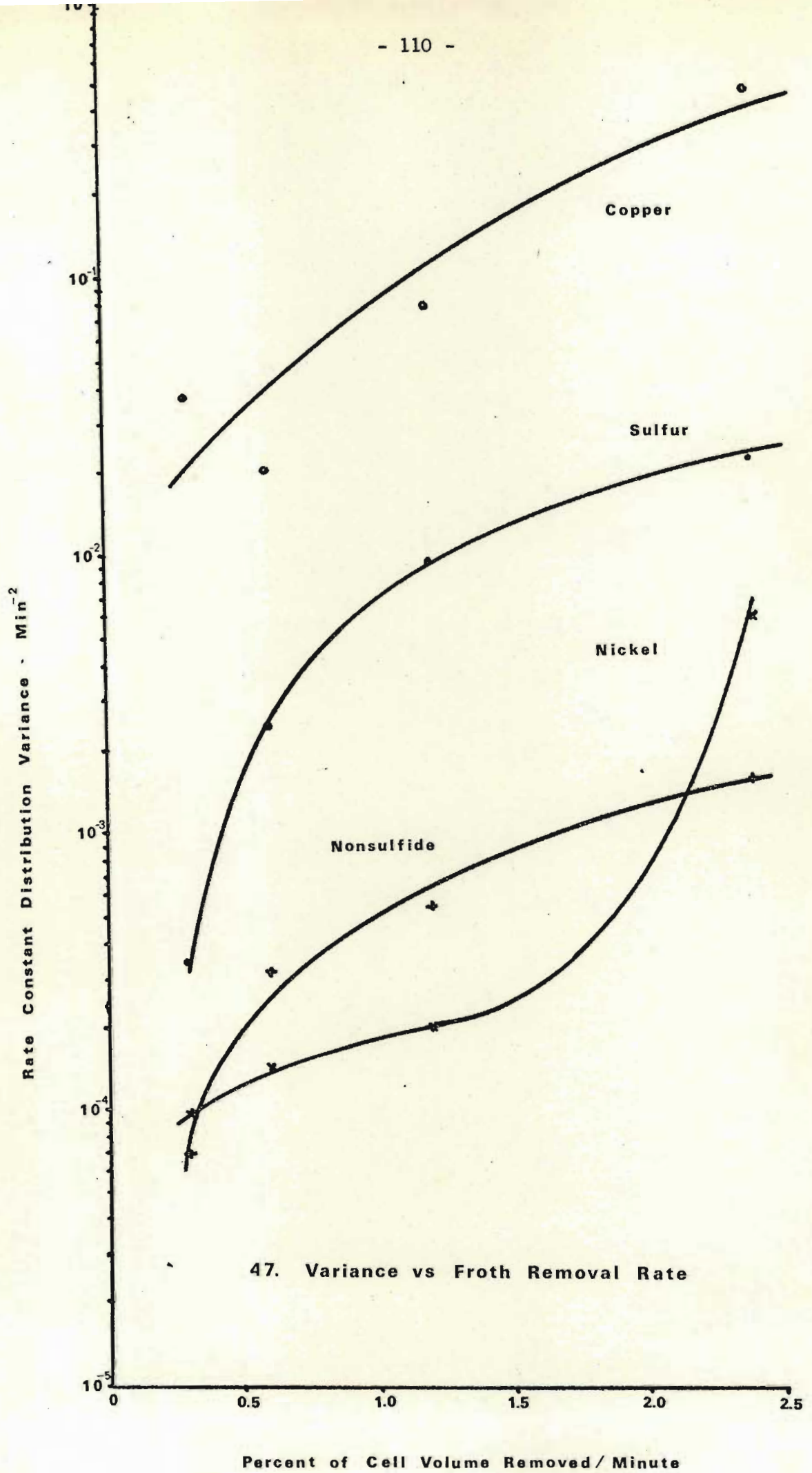


TABLE XI. FROTH REMOVAL RATE DATA - ROUGHER CONCENTRATE BATCH TESTS

Species and Test Code	% of Cell Volume removed as Concentrate/min	Rate Plus Nonfloating Model			Gamma Function Model		
		k	C <sub>n</sub>	SSQS	k	σ <sup>2</sup>	SSQS
Sulfur							
311--170311320	0,3	0,06727	0,370	0,01445	0,03371	0,3403E-3	0,1425E-2
" " 1610	0,6	0,06637	0,181	0,7216E-3	0,06120	0,2426E-2	0,2280E-3
" " 1500	1,2	0,09720	0,169	0,4244E-2	0,1040	0,9413E-2	0,9667E-3
" " 1715	2,4	0,1359	0,183	0,2524E-2	0,1476	0,02100	0,2499E-3
Nonsulfide							
321--170311320	0,3	0,01859	0,799	0,2304E-4	0,3744E-2	0,6806E-4	0,1700E-4
" " 1610	0,6	0,1507	0,894	0,3692E-2	0,6910E-2	0,3134E-3	0,3816E-5
" " 1500	1,2	0,03702	0,704	0,1162E-3	0,01164	0,5330E-3	0,4031E-4
" " 1715	2,4	0,05050	0,622	0,1469E-3	0,02154	0,1480E-2	0,4734E-4
Copper							
331--170311320	0,3	0,1100	0,089	0,01827	0,1919	0,03684	0,3144E-2
" " 1610	0,6	0,1146	0,022	0,01414	0,1855	0,02026	0,6800E-3
" " 1500	1,2	0,1309	0,000	0,04408	0,3320	0,07644	0,1244E-3
" " 1715	2,4	0,3177	0,095	0,3719E-2	0,6253	0,4515	0,3660E-3
Nickel							
341--170311320	0,3	0,06210	0,305	0,1200	0,03461	0,8854E-4	0,06455
" " 1610	0,6	0,07922	0,139	0,05169	0,05878	0,1503E-3	0,02395
" " 1500	1,2	0,09344	0,064	0,9415E-2	0,07765	0,1986E-3	0,3225E-2
" " 1715	2,4	0,1327	0,056	0,1074E-2	0,1401	0,5779E-2	0,4249E-3



TABLE XII - MODEL COMPARISON

PREDICTION OF CONTINUOUS OPERATION

Material	Mean Error in Predicted Concentration for Model Form	
	Rate plus Non-floating	Gamma Function
Sulphur - Rougher	,19	,17
" Cleaner	,135	,136
Copper - Rougher	,13	,07
" Cleaner	,11	,13
Nickel - Rougher	,11	,07
" Cleaner	,12	,13
Non-sulphide - Rougher	,005	,0045
" Cleaner	,10	,10
Total - Rougher	,169	,137
" Cleaner	,151	,160

TABLE XIII - MODEL COMPARISON  
PREDICTIONS FOR REFLOATED MATERIAL

Species	Mean Error in Predicted Concentration for Model Form		
	Rate plus Non-floating	Gamma Function	
		$(\bar{k}, \sigma^2)$	$(\bar{k}, \sigma^2, C_n)$
Sulphur	0,18	0,16	0,148
Copper	0,23	0,23	0,191
Nickel	0,22	0,21	0,175
Total	0,20	0,19	0,16

TABLE XIV - MODEL COMPARISON

BATCH TEST DESCRIPTION

Species	Mean Error in Fitted Concentration for Model Form	
	Rate plus Non-floating	Gamma Function
Sulphur - Rougher	,04	,03
" - Cleaner	,009	,01
Copper - Rougher	,03	,02
" Cleaner	,01	,005
Nickel - Rougher	,03	,04
" Cleaner	,009	,01
Non-sulphide - Rougher	,004	,004
" Cleaner	,005	,009
Total -	,0074	,0067

All batch tests from continuous prediction and prediction of  
refloated material are used.



V. USE OF THE GAMMA FUNCTION MODEL AS A CIRCUIT  
DESIGN AID.

The mathematical technique developed by Woodburn (refer to section VIII) allows a rapid method of flotation circuit design analysis based on what is the most descriptive of the first-order irreversible models investigated; the continuous distribution gamma function model. In the previous chapters, the gamma function model has been proved accurate in the description of batch flotation and its predictive capabilities have been proved reasonable considering the difficulties of the froth removal rate and of the machine dependability of parameters. In continuation, the gamma function model's assets and liabilities can be further illustrated by a circuit design analysis of the Rustenburg ores.

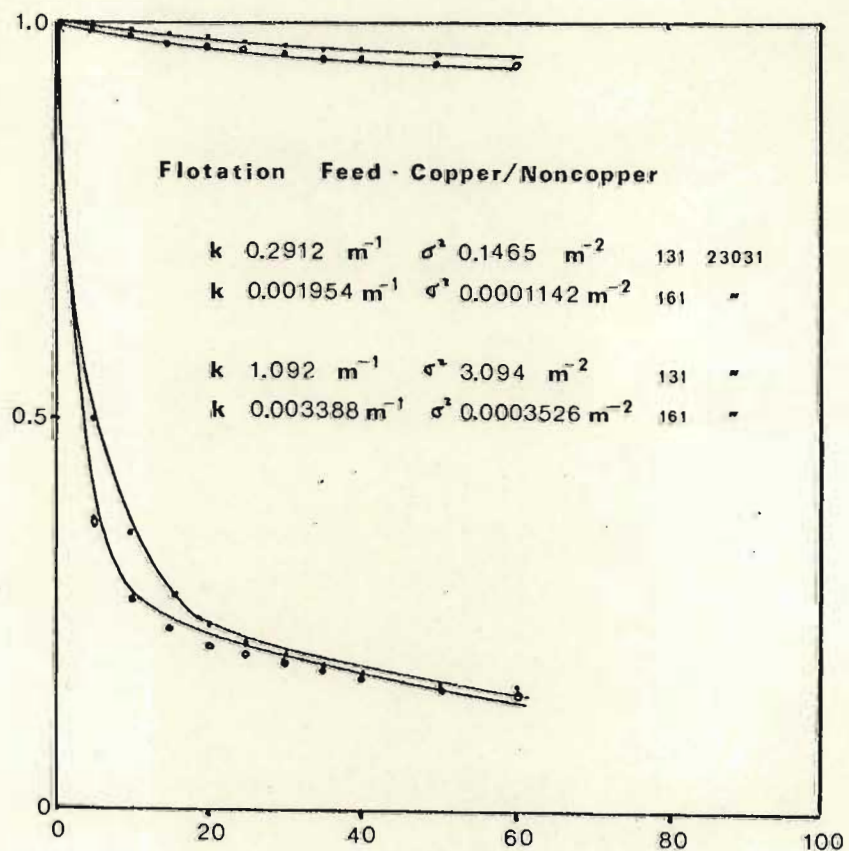
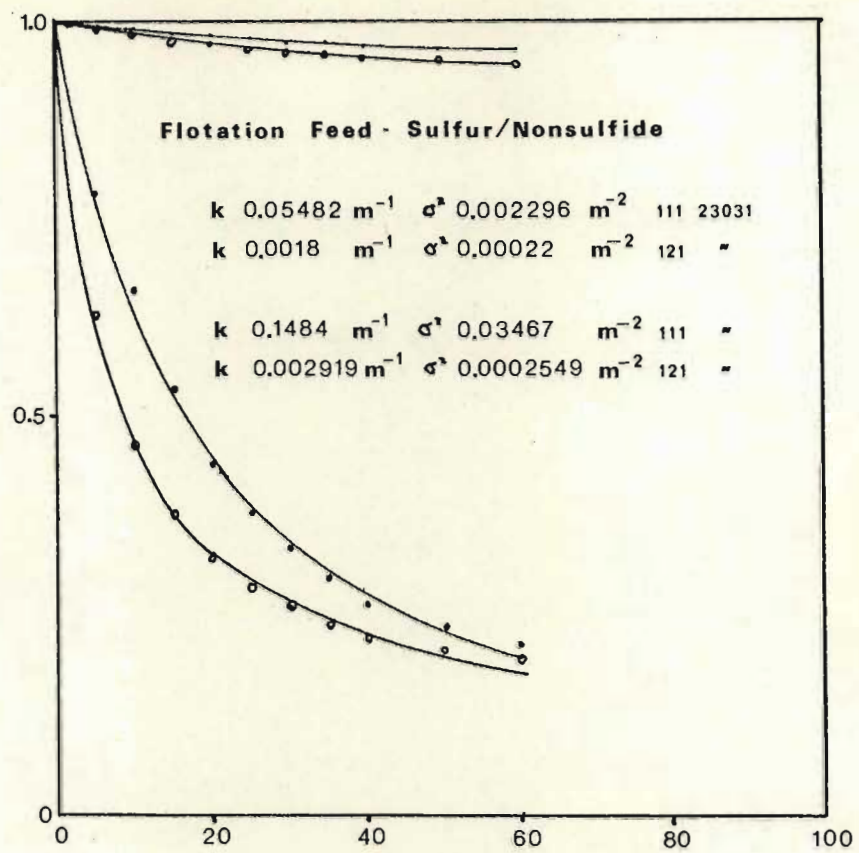
The true test of a model's predictive capability is the analysis of plant scale flotation in an industrial circuit by means of batch tests on the plant feed stream. Two flotation tests on the plant feed stream were performed, their concentration versus time curves are shown in figures 48 and 49. The froth removal rates were not measured, however, it is assumed that the faster apparent flotation rate is more representative of the plant performance and simulations of the various plant circuits are based on the regressed parameters of test 2. The batch tests and the sampling campaigns on the various circuits described later were not performed at the same time. In general operation of the plant, results are fairly steady showing that the feed material does not alter radically over long periods thus the batch test was performed on a material which is reasonably representative of the feed to the various circuits.

### Simulation of Various Hypothetical Circuits.

In the simulation of the following circuits it is necessary to provide some basis of comparison, this is best done from the point of view of plant utilization, i.e. maximum metallurgical results from a limited capital investment. It is convenient to assume a plant availability of twenty-five equal volumed units which may be arranged in any combination to produce the desired circuit. Four basic circuits are of interest: the rougher-scavenger circuit, the rougher-cleaner-recleaner circuit, the rougher-cleaner circuit and the rougher-cleaner-cleaner middlings circuit. Flow diagrams are provided in Figure 50. In the generation of the grade-recovery curves of Figure 51, an increasing number of cells were used to produce the final concentrate stream. In the case of the rougher-cleaner-recleaner circuit and the rougher-cleaner-cleaner middlings circuit, the number of cleaner and cleaner middlings cells, respectively, were kept constant at four units.

The simulations shown have many of the trends that experience would dictate. Firstly, recovery increases as grade is dropped in the simple rougher-scavenger circuit. Secondly, in all those circuits which utilize cleaner capacity a sharp drop in the recovery-grade curve is experienced when losses due to the shortening of the rougher bank override increases in recovery due to the production of a lower grade concentrate. Thirdly, the faster floating copper material is suited to a cleaner operation while the slower floating nickel material is not. Slow floating materials are both difficult to recover and difficult to produce at good grades; the nearer the flotation rate of the valuable species gets to the rate of flotation of the gangue species, the less advantageous a cleaning operation becomes.

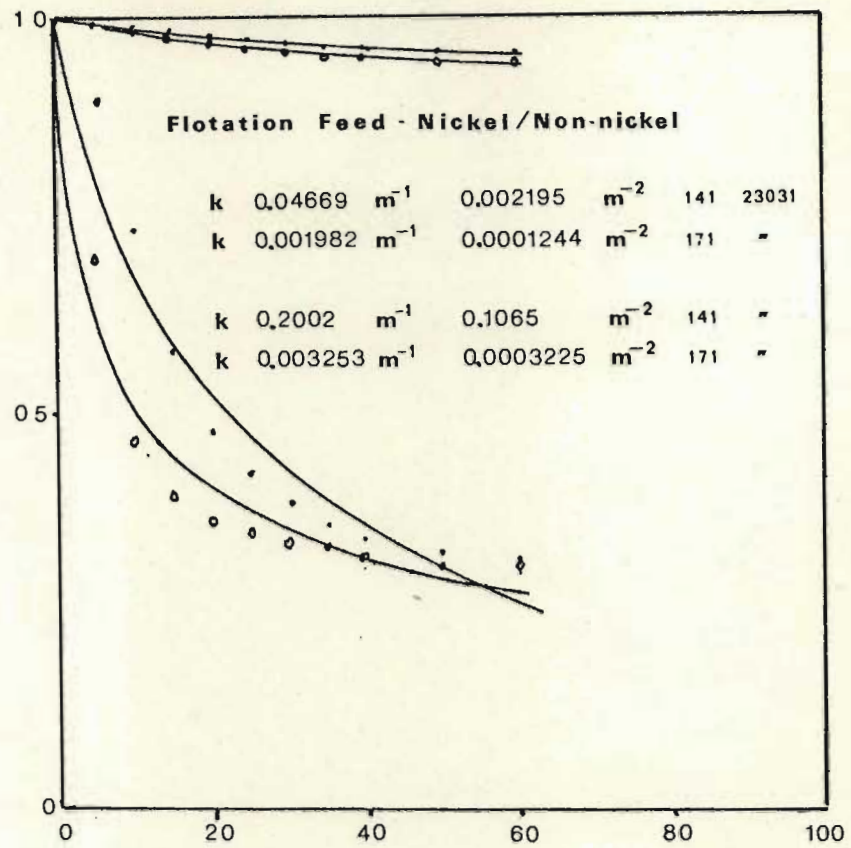




#### 48. Batch Tests - Gamma Function Model

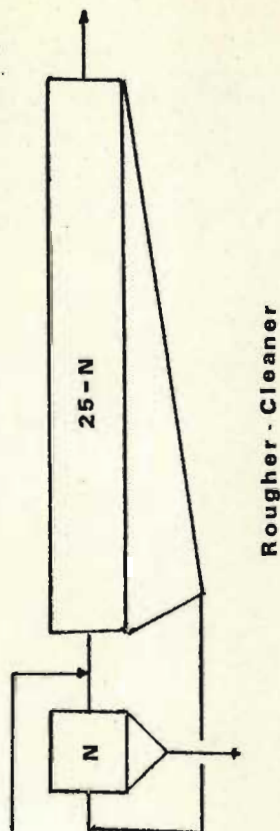
Fraction in Tailings vs Time - mins



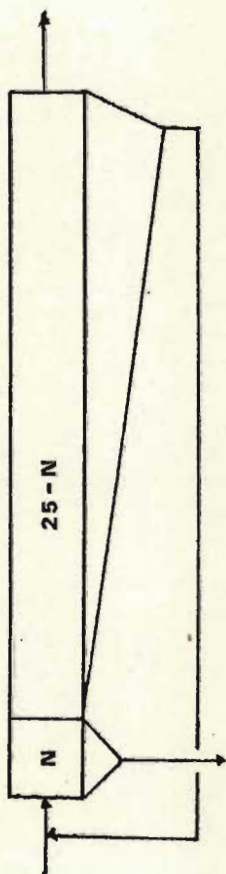


**49. Batch Tests - Gamma Function Model**

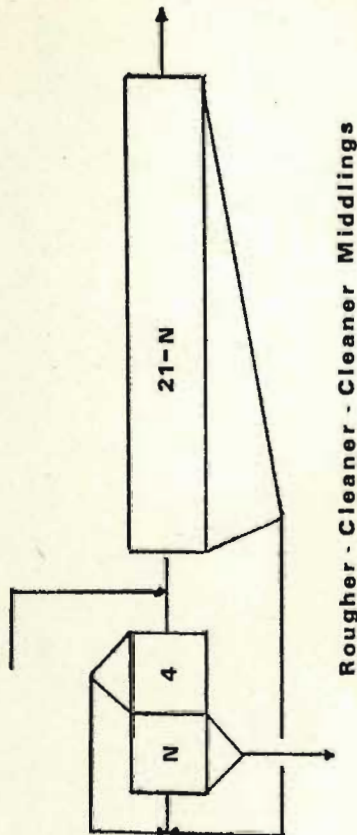
**Fraction in Tailings vs Time - mins**



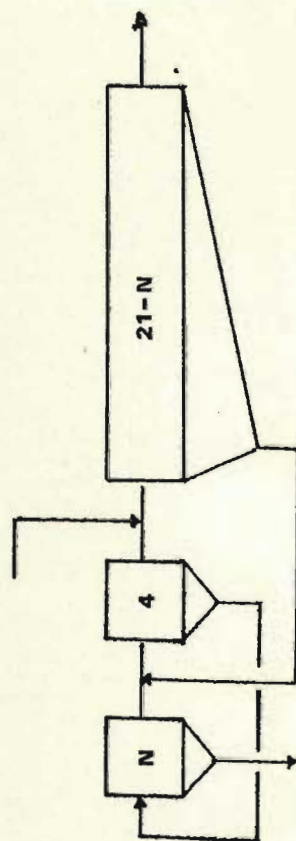
Rougher - Cleaner



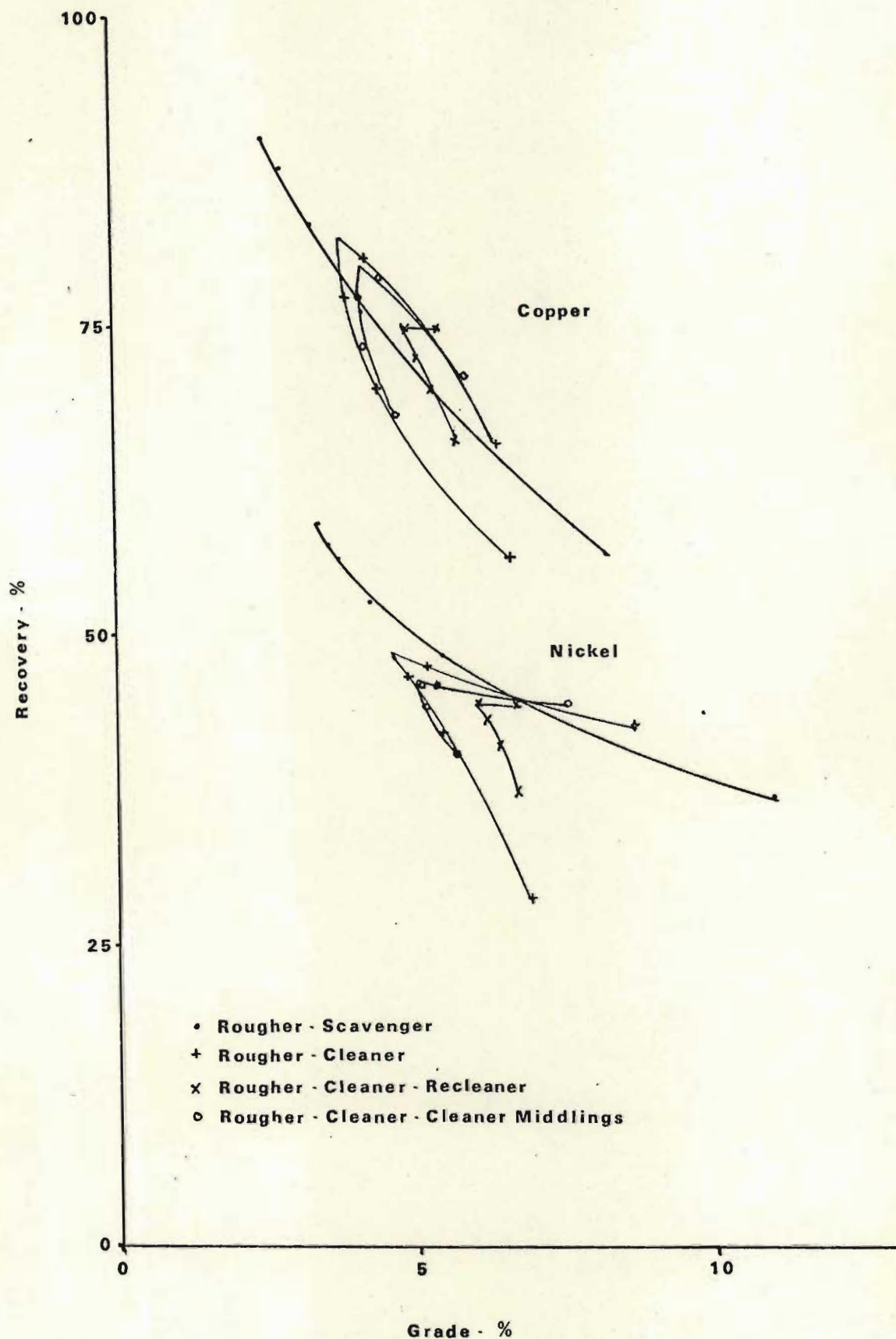
Rougher - Scavenger



Rougher - Cleaner - Cleaner Middlings



Rougher - Cleaner - Recleaner



51. Simulated Grade-Recovery Curves



### Comparison of Actual Circuit Data with Simulations

Although the gamma function model may be capable of predicting operational trends for various circuit configurations as it has done above, it is also of value to determine the range of error in a circuit simulation. Several flotation cell configurations were tested at Rustenburg over a period of several months. The flotation section tested at Rustenburg may be simplified into a forty unit network; twenty of these units have four times the effective volume of the remaining twenty. The smaller volumed units serve as cleaners. Use of Woodburn's technique (refer to Section VIII) then involves the simulation of flotation in these forty units when the units are arranged so as to produce a rougher-scavenger, a rougher-cleaner, a rougher-cleaner-recleaner, a rougher-cleaner-cleaner middlings, and a rougher-scavenger-cleaner-cleaner middlings circuit. For the simulation, a ratio of concentrate volume to tailings volume was assumed for each cell. The assumed value for rougher and scavenger cells is 0,011 and for cleaner, recleaner or cleaner middling cells is 0,025. Figure 52 illustrates the circuits as operated, smaller cells are indicated as such. Data for the rougher-scavenger circuit is available on a periodic basis throughout the test period and sulphur, copper and nickel analyses are available. All other data represents test averages; copper and nickel analyses are available only for one case of the rougher-cleaner-cleaner middlings circuit. Simulated results and actual assays are presented in Tables XV and XVI. Table XVII contains general operational data for the test circuits.

Generally the model performs poorly; the batch test kinetic parameters predict higher tailings losses and higher final concentrate grades in the majority of cases. The only exception being the rougher-cleaner circuit sulphide simulation

where the model predictions compare very well with the operational results. It is worthwhile to note that the model's simulated values in the rougher-scavenger circuit lie outside the range of one and one-half standard deviations from the average value as assayed; an exception is the percentage of nickel in the final concentrate. The two probable sources of error have already been discussed, namely, the dependence of the batch kinetic parameters upon froth removal rate and machine design. As is illustrated in the section on the applicability of the four flotation models, an increase in froth removal rate will increase the apparent rate of flotation of both the gangue and the valued species. It is possible that the froth removal rate in the batch test used for simulation was lower than that which corresponds to plant operation since an increased froth removal rate would lower both tailings grades and concentrate grades.

### Conclusions

The requirement of a reasonably accurate production and design tool has not been completely met. Of the four models investigated, the gamma function first-order irreversible model has proved most acceptable in the conditions tested. Use of the gamma function model as a production and design aid is of some value, however, considerable caution should be exercised in the use of simulated results.

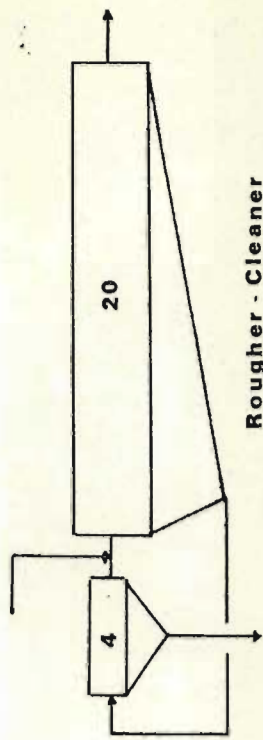
The major problems associated with the use of the gamma function model in its present form are

- 1) some species may be particularly susceptible to the flotation action in different machines, thus the measured rate parameters are machine dependent ,
- 2) the froth phase is an important and integral part of flotation, its removal rate does in fact determine the apparent rate of species flotation.

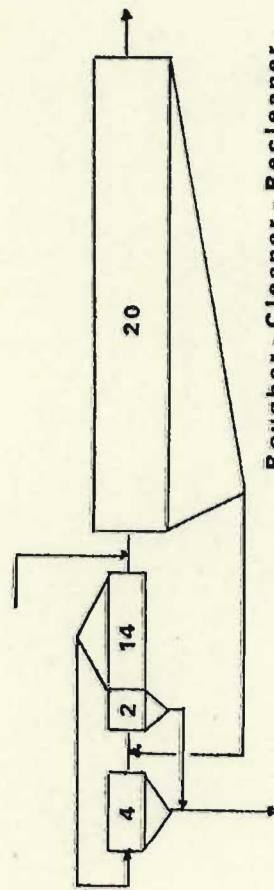




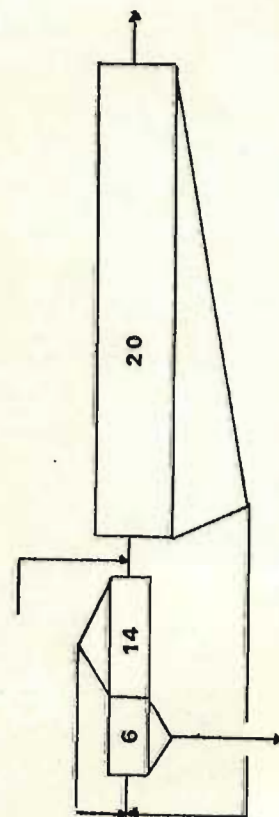
Rougher - Scavenger



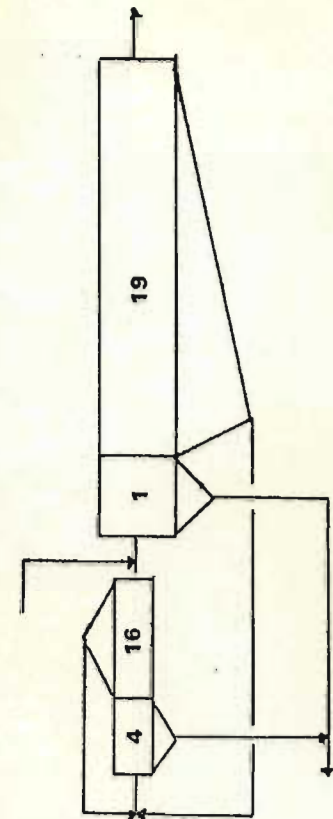
Rougher - Cleaner



Rougher - Cleaner - Recleaner



Rougher - Cleaner - Cleaner Middlings



Rougher - Scavenger - Cleaner - Cleaner Middlings



TABLE XV. DATA SUMMARY FOR ROUGHER-SCAVENGER CIRCUIT

Time	Measured Feed Assay			Tailings Assays						Concentrate Assays					
				Measured Value			Model Value			Measured Value			Model Value		
	% Cu	% Ni	% S	% Cu	% Ni	% S	% Cu	% Ni	% S	% Cu	% Ni	% S	% Cu	% Ni	% S
730	0,065	0,164	0,38	0,016	0,064	0,14	0,024	0,097	0,23	3,05	5,73	15,0	3,77	6,27	14,8
800			1,08	0,016	0,059	0,14			0,65	3,33	4,85	13,3			33,3
815	0,077	0,264	0,61			0,14	0,028	0,158	0,37			12,1	4,48	9,81	22,0
830			0,47	0,018	0,080	0,13			0,28	3,63	6,81	17,5			17,9
845	0,071	0,182	0,45			0,13	0,026	0,108	0,27			13,3	4,13	6,94	17,1
900			0,46	0,015	0,078	0,15			0,28	4,34	6,48	15,1			17,8
915	0,068	0,180	0,42			0,15	0,025	0,108	0,26			18,6	4,07	7,02	16,4
930			0,48	0,015	0,072	0,17			0,30			12,2			18,6
945	0,075	0,195	0,44			0,15	0,027	0,116	0,26	3,34	7,05	13,8	4,35	7,40	16,8
1000			0,42	0,021	0,090	0,18			0,25			14,5			15,9
1015	0,063	0,185	0,40			0,19	0,022	0,108	0,23	3,89	5,04	13,1	3,48	6,73	15,0
1030			0,46	0,016	0,073	0,13			0,27			12,5			17,1
1045	0,073	0,166	0,43			0,14	0,026	0,098	0,25	3,75	6,26	16,4	4,11	6,21	16,2
1100			0,43	0,018	0,077	0,15			0,26			14,2			16,4
1115	0,069	0,168	0,42			0,13	0,025	0,099	0,25	3,68	4,62	13,9	3,95	6,35	16,0
1130			0,51	0,018	0,074	0,15			0,30			18,4			18,6
1145	0,078	0,187	0,48			0,13	0,028	0,110	0,28	3,72	7,19	17,2	4,36	6,91	17,7
1200			0,50	0,014	0,074	0,14			0,30			13,8			18,6
1215	0,096	0,174	0,41			0,15	0,034	0,102	0,24	3,51	6,26	15,7	5,35	6,50	15,6
1235			0,46	0,013	0,094	0,19			0,27	3,10	5,85	14,9			17,1
1250	0,074	0,214	0,46			0,14	0,027	0,127	0,27			14,9	4,25	8,00	17,4

TABLE XV. (Continued)

Time	Measured Feed Assay			Tailings Assays						Concentrate Assays					
				Measured Value			Model Value			Measured Value			Model Value		
	% Cu	% Ni	% S	% Cu	% Ni	% S	% Cu	% Ni	% S	% Cu	% Ni	% S	% Cu	% Ni	% S
315			0,47	0,010	0,114	0,18			0,28	3,38	7,51	18,6			17,6
345	0,076	0,187	0,56	0,008	0,099	0,16	0,026	0,107	0,32	3,16	10,95	16,9	4,06	6,62	19,5
400	0,074	0,135	0,43			0,19	0,027	0,079	0,25			13,5	4,18	5,12	16,2
415			0,55	0,008	0,099	0,20			0,33	3,35	4,50	11,1			20,0
430	0,084	0,198	0,46			0,18	0,030	0,116	0,27			11,7	4,68	7,29	17,1
445			0,44			0,15			0,26	3,36	4,47	12,2			16,2
500	0,093	0,219	0,45	0,006	0,090	0,16	0,032	0,127	0,26			12,3	5,06	7,86	16,6
515			0,56			0,16			0,33	3,47	6,24	15,8			20,2
530	0,098	0,228	0,52	0,024	0,099	0,17	0,035	0,133	0,30			12,0	5,37	8,24	18,8
545			0,52			0,19			0,31	2,59	4,97	13,0			19,0
AGE	0,077	0,190	0,475	0,015	0,084	0,16	0,028	0,112	0,29	3,45	6,16	14,4	4,35	7,08	18,0
DEV.	0,011	0,030	0,146	0,005	0,015	0,02	0,004	0,018	0,08	0,39	1,58	2,2	0,53	1,06	3,3



TABLE XVI. DATA SUMMARY FOR CIRCUIT ANALYSIS

Circuit Description	Feed Assay (measured)	Tails Assay		Concentrate Assay		Test Length (hours)
		(measured)	(Model)	(measured)	(model)	
Rougher-Scavenger-Cleaner						
Cleaner-Middling (% S)	0,42	0,14	0,16	7,4	10,6	12
Rougher-Cleaner (% S)	0,43	0,10	0,10	13,4	12,3	8,3
Rougher-Cleaner- Recleaner (% S)	0,47	0,18	0,25	12,8	18,6	8
Rougher-Cleaner- Cleaner-Middling						
TEST 1 (% S)	0,41	0,15	0,20	8,6	14,5	12
TEST 2 (% S)	0,43	0,15	0,25	7,0	18,5	9,8
TEST 3 (% S)	0,43	0,16	0,20	7,0	14,3	12
(% Cu)	0,077	0,0134	0,0143	2,51	3,76	12
(% Ni)	0,191	0,092	0,103	3,42	5,45	12
TEST 4 (% S)	0,44	0,14	0,18	7,1	12,3	11,5



TABLE XVII.

Test Circuit	General Remarks
Rougher-Scavenger-Cleaner-Cleaner Middlings	Average particle size slightly finer than normal. Residence time in rougher-scavenger nearly double of normal operation.
Rougher-Cleaner	Average particle size normal, residence time in rougher also normal.
Rougher-Cleaner-Recleaner	Particle size and residence time in rougher normal.
Rougher-Cleaner-Cleaner Middling	Test 1, Very fine average particle size; residence time in rougher slightly longer than normal. Test 2, Normal operation. Test 3, Normal operation. Test 4, Particle size slightly coarser than normal; residence time in rougher bank nearly double normal.
Rougher-Scavenger	Normal operation.

### Mathematical Formulation of the Models

The mathematical description of each model is discussed and the equations necessary to handle various predictions are outlined in detail in the following sections in order to clarify the results of the investigation. Each of the investigated models is discussed separately.

#### The Single Rate Model

The single rate model may be developed for the batch case as follows

$$dc/dt = -kC$$

hence through integration and initial conditions

$$C = C_0 e^{-kt}$$

The addition of the 'perfect' mixing residence time distribution below

$$E(t) = \frac{1}{T} e^{-t/T}$$

leads to the equation for continuous operation of a single cell

$$C = C_0 \int_0^{\infty} e^{-kt} \frac{1}{T} e^{-t/T} dt$$

$$C = \frac{C_0}{(1 + kT)}$$

By repeated application of the above equation, the equation for a series of 'n' equal volumed units may be developed.

$$C = \frac{C_0}{(1 + kT)^n}$$

As the single rate model describes the flotation of any mineral species by a single first-order rate constant, the same equations and rate constant apply to the species at any point within a flotation network.

#### The Rate plus Non-floating Model

Mathematically this model is a simple extension of the single rate model into a general discrete distribution model as follows :

$$dC_i/dt = -k_i t$$

thus for batch operation

$$C = \sum_{i=1}^n C_i e^{-k_i t} \quad ; \quad \sum_{i=1}^n C_i = C_0$$

and for continuous operation

$$C = \sum_{i=1}^n \frac{C_i}{(1 + k_i T)^n}$$

For the forms used in fitting data at Rustenburg where only two fractions were assumed,  $C_n$ , the fraction which is unfloatable and the remaining material which floats according



to a single rate constant, the equation for a batch test becomes

$$C = (C_o - C_n) e^{-kt} + C_n$$

and for continuous operation

$$C = [(C_o - C_n) / (1 + kT)^n] + C_n$$

In the application to various circuits with the rate plus non-floating model, the concentrate from any cell contains no material with a zero rate constant, in fact all the concentrate material must behave as floatable material with a rate constant of 'k'. Thus in the particular application of predicting rougher concentrate flotation from the rougher feed batch test, the rougher feed batch test is described by the equation

$$C = (C_o - C_n) e^{-kt} + C_n$$

and the rougher concentrate batch test is described by

$$C = C_o e^{-kt}$$

#### The Gamma Function Model

Woodburn and Loveday<sup>38</sup> have assumed a form of the gamma function to describe a distribution of first-order rate constants for any mineral species; the assumed distribution is

as follows

$$W(k) = \frac{(\bar{k}/\sigma^2)^{(\bar{k}^2/\sigma^2)} k^{(\bar{k}^2/\sigma^2 - 1)} e^{-k\bar{k}/\sigma^2}}{\Gamma(\bar{k}^2/\sigma^2)}$$

since

$$\int_0^{\infty} k^{(\bar{k}^2/\sigma^2 - 1)} e^{-k\bar{k}/\sigma^2} dk = \frac{\Gamma(\bar{k}^2/\sigma^2)}{(\bar{k}/\sigma^2)^{(\bar{k}^2/\sigma^2)}}$$

$$\int_0^{\infty} W(k) dk = 1$$

for all  $\bar{k} > 0$  and  $\sigma^2 > 0$ .

By simply applying the batch equation for the discrete distribution model, the equation for the gamma function model may be developed

$$C = \sum_{i=1}^n C_i e^{-k_i t}$$

$$C = \sum_{i=1}^n W\left(\frac{k_i + k_{i+1}}{2}\right) \exp\left[\left(\frac{k_i + k_{i+1}}{-2}\right)t\right]$$

and letting  $n \rightarrow \infty$

$$C = \int_0^{\infty} W(k) e^{-kt} dk$$

or

$$C = \int_0^{\infty} \frac{(\bar{k}/\sigma^2)^{(\bar{k}^2/\sigma^2)} k^{(\bar{k}^2/\sigma^2 - 1)} e^{-k\bar{k}/\sigma^2 - kt}}{\Gamma(\bar{k}^2/\sigma^2)} dk$$

and upon collection of terms and integration

$$C = \frac{(\bar{k}/\sigma^2)^{(\bar{k}^2/\sigma^2)}}{\Gamma(\bar{k}^2/\sigma^2)} \int_0^{\infty} k^{(\bar{k}^2/\sigma^2 - 1)} e^{-k(\bar{k}/\sigma^2 + t)} dk$$

$$C = \left( \frac{\bar{k}/\sigma^2}{\bar{k}/\sigma^2 + t} \right)^{(\bar{k}^2/\sigma^2)}$$

In the continuous operation of a single cell,

$$C = \int_0^{\infty} \left( \frac{\bar{k}/\sigma^2}{\bar{k}/\sigma^2 + t} \right)^{(\bar{k}^2/\sigma^2)} \frac{1}{\tau} e^{-t/\tau} dt$$

however, since  $\bar{k}$  and  $\sigma^2$  define only the distribution of rate constants in the feed and this distribution will alter throughout the circuit, repeated application of the above equation is incorrect. Loveday has developed a form of the above equation which may be used to predict operation of a series of cells. Using the 'perfectly mixed' residence time distribution for a single unit and Laplace transforms, the residence time distribution for a series of cells is

$$E(t) = \frac{1}{\tau(n-1)!} \left( \frac{t}{\tau} \right)^{n-1} e^{-t/\tau}$$



where  $n$  is the number of equal residence time units with a mean holding time of  $T$ .

Thus for continuous operation of a series of cells

$$C = \int_0^{\infty} \int_0^{\infty} W(k) e^{-kt} E(t) dk dt$$

$$C = \frac{1}{T^n (n-1)!} \int_0^{\infty} \left( \frac{k/\sigma^2}{k/\sigma^2 + t} \right)^{(k^2/\sigma^2)} t^{n-1} e^{-t/T} dt$$

A general mathematical formulation of flotation networks has been put forward by Dieft<sup>39</sup>. Woodburn has expanded Dieft's work and developed the general network equations for the gamma function or any distributed parameter model.

With Professor E.T. Woodburn's permission, I am publishing his work here.

Consider a tank 'j' as shown in figure 53. Let  $d\alpha(k)dk$  be the mass flow of tailings which has an apparent first-order rate constant in the region  $k$  to  $k + dk$ . Using the notation that the fraction of the tailings stream from cell  $j$  which is feed to cell  $i$  equals  $a_{ij}$ ; and similarly the fraction of the concentrate stream from cell  $j$  which feeds cell  $i$  equal to  $b_{ij}$ . The feed to any cell  $j$

becomes

$$\sum_{\substack{k=1 \\ k \neq i,j}}^{k=N} (a_{jk} + q_k b_{jk}) d\alpha_k + dF_j = (1 + q_j) d\alpha_j$$

Defining  $a_{jj} = -1$  and  $b_{jj} = -1$  then

$$\sum_{k=1}^{k=N} (a_{jk} + b_{jk} q_k) d\alpha_k = -dF_j$$

and in matrix notation

$$\underline{C} d\alpha = -dF$$

or

$$d\alpha = -\underline{C}^{-1}$$

If all mass flows are normalized and the rate constant distribution of the feed stream is  $W(k)$ , then

$$dF = \underline{\epsilon} W(k) dk$$

where  $\underline{\epsilon}$  is a column matrix containing the fraction of fresh feed to each cell, and

$$\alpha = - \int_0^{\infty} \underline{C}^{-1} \underline{\epsilon} W(k) dk$$

The same form of the above formula may be used to generate a series of residence times for each of the flotation

units in a circuit. It is convenient and reasonably practical to work in terms of the volume split within a cell. Thus following the same argument as for the solids but with a constant value for  $g_i$ ; now the ratio of concentrate volume to tailings volume, the mean residence time in each unit may be calculated directly.

$$\underline{Q}_v = - \underline{C}^{-1} \underline{E} F_v$$

The fractional recovery of any species is defined as the total mass flowrate of that species in all concentrate streams which are not recycled, for a fresh feed of unity for that species. Thus the partial fractional recovery associated with the production of concentrate from cell  $j$  is

$$r_j = - \sum_{i=1}^{i=N} b_{ij}$$

one minus the sum of all fractions recycled (remembering that  $b_{jj} = -1$ ).

The concentrate flowrate of a species is

$$\underline{g} d\underline{x} = - \underline{g} \underline{C}^{-1} d\underline{F}$$

where  $\underline{g}$  is a matrix such that

$$g_{jj} = kT_j$$

(where  $T_j$  is mean residence time in cell  $j$ )

$$g_{ij} = 0 \quad (i \neq j)$$



Therefore the fractional recovery of any given mineral species is defined by

$$R = - \int_0^{\infty} \underline{r}^T \underline{g} \underline{C}^{-1} dF$$

or

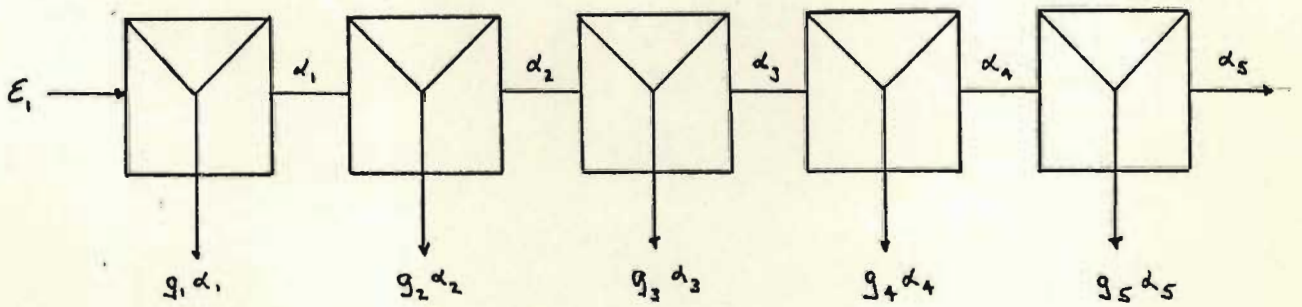
$$R = - \int_0^{\infty} \underline{r}^T \underline{g} \underline{C}^{-1} \underline{\epsilon} W(k) dk$$

The rate constant distribution in any concentrate stream is defined by

$$\bar{k} = \frac{\int_0^{\infty} k \underline{r}^T \underline{g} \underline{C}^{-1} \underline{\epsilon} W(k) dk}{\int_0^{\infty} \underline{r}^T \underline{g} \underline{C}^{-1} \underline{\epsilon} W(k) dk}$$

$$\sigma^2 = \frac{\int_0^{\infty} (k - \bar{k})^2 \underline{r}^T \underline{g} \underline{C}^{-1} \underline{\epsilon} W(k) dk}{\int_0^{\infty} \underline{r}^T \underline{g} \underline{C}^{-1} \underline{\epsilon} W(k) dk}$$

In example of the above formulation, consider a simple bank of five continuously operating flotation cells with the total feed entering the first cell and with no concentrate streams being recycled.



For this case then  $\epsilon_1 = 1$ ,  $\epsilon_i = 0$  for  $1 < i \leq 5$ . By definition  $a_{jj} = b_{jj} = -1$  and  $g_i = kT_i$  if a perfectly mixed unit is assumed, thus  $C_{jj} = -(1 + g_j)$ .

Due to the cell configuration  $a_{j+1,j} = 1$  for  $1 \leq j < 5$ ;  $a_{jk} = 0$  for  $1 \leq k \leq 5$ ,  $1 \leq j \leq 5$  but  $j \neq k, k+1$ ; and  $b_{jk} = 0$   $1 \leq k \leq 5$ ,  $1 \leq j \leq 5$  but  $j \neq k$ .

In this case then the configuration matrix will be

$$C = \begin{bmatrix} C_{11} & 0 & 0 & 0 & 0 \\ 1 & C_{22} & 0 & 0 & 0 \\ 0 & 1 & C_{33} & 0 & 0 \\ 0 & 0 & 1 & C_{44} & 0 \\ 0 & 0 & 0 & 1 & C_{55} \end{bmatrix}$$

Its inverse will be

$$\underline{C}^{-1} = \begin{bmatrix} C_{11}^{-1} & 0 & 0 & 0 & 0 \\ -C_{11}^{-2} & C_{11}^{-1} & 0 & 0 & 0 \\ C_{11}^{-3} & -C_{11}^{-2} & C_{11}^{-1} & 0 & 0 \\ -C_{11}^{-4} & C_{11}^{-3} & -C_{11}^{-2} & C_{11}^{-1} & 0 \\ C_{11}^{-5} & -C_{11}^{-4} & C_{11}^{-3} & -C_{11}^{-2} & C_{11}^{-1} \end{bmatrix}$$

The general expression for  $\underline{\alpha}$  the tailings flow is,

$$\underline{\alpha} = \int_0^{\infty} \underline{C}^{-1} \underline{\varepsilon} W(k) dk$$



thus we have the following matrix expression,

$$\begin{bmatrix} \alpha_1 \\ \alpha_2 \\ \alpha_3 \\ \alpha_4 \\ \alpha_5 \end{bmatrix} = \int_0^{\infty} \begin{bmatrix} \frac{1}{1+q_1} & 0 & 0 & 0 & 0 \\ \frac{1}{(1+q_1)^2} & \frac{1}{1+q_1} & 0 & 0 & 0 \\ \frac{1}{(1+q_1)^3} & \frac{1}{(1+q_1)^2} & \frac{1}{1+q_1} & 0 & 0 \\ \frac{1}{(1+q_1)^4} & \frac{1}{(1+q_1)^3} & \frac{1}{(1+q_1)^2} & \frac{1}{1+q_1} & 0 \\ \frac{1}{(1+q_1)^5} & \frac{1}{(1+q_1)^4} & \frac{1}{(1+q_1)^3} & \frac{1}{(1+q_1)^2} & \frac{1}{(1+q_1)} \end{bmatrix} \begin{bmatrix} 1 \\ 0 \\ 0 \\ 0 \\ 0 \end{bmatrix} W(k) dk$$

or

$$\alpha_j = \int_0^{\infty} \frac{W(k) dk}{(1+q_1)^j}$$

The above integrals for the calculation of species recovery, concentrate rate constant distribution and fractional tailings flow, when the modified gamma function distribution of rate constants is used, do not have an analytical solution. However, use of generalized Laguerre polynomials and Gauss quadrature techniques simplifies

digital integration to a realizable form. The general integral of the form,

$$\int_0^{\infty} F(x) x^{\beta} e^{-x} dx$$

may be digitally solved by generation of the correct roots and weights for the following expression.

$$\int_0^{\infty} F(x) x^{\beta} e^{-x} dx = \sum_{i=1}^{i=NN} f(x_i) w_i$$

The author is indebted to Dr. I. Kropholler of Loughborough University for the contribution of a Fortran program for calculation of the necessary roots and weights which are both a function of  $\beta$ .

#### The Two-phase Model.

This model, the only froth incorporating model to be discussed, was examined in 1966 by Harris and Rimmer<sup>40</sup>. The equations are merely presented here and their application discussed.

For the batch or semi-batch case,

$$C = \frac{C_0}{E - F} \left[ E e^{Ft} - F e^{Et} \right]$$

where E and F are the roots for 'Z' to the following equation

$$(Z + a)(Z + b + Qf/Vf) = ab$$

'a' and 'b' being the rate constant for mass transfer from pulp to froth and froth to pulp respectively. The term  $Qf/Vf$  is assumed constant with time, and represents the froth flowrate divided by the froth volume.

Since  $Qf/Vf$  was not to be function of time, calculation of concentration with time was done in the following manner for batch tests,

$$C(t_i) = \frac{C(t_{i-1})}{E - F} \left[ E e^{Ft} - F e^{Et} \right]$$

where 'i' corresponds to the time interval, i.e. 0 - 5 mins., 5 - 10 mins. etc.  $Qf/Vf$  is measured over this interval and assumed constant.

The equations developed by Harris and Rimmer for continuous operation are not presented here because the model showed very poor results in batch testing analysis and no further investigation was thought necessary at this time. Data for a single bank of continuously operated Fagergren 40 cu.ft. cells in which  $Qf/Vf$  were measured is available however.

#### Parameter Estimation Techniques

Many methods are available for the best estimation of parameters for any model; broadly speaking, they may be divided into three general categories; derivative, non-derivative and search techniques. The methods of Fletcher



and Powell<sup>41,42</sup> Rosenbrock<sup>43</sup> Hooke and Jeeves<sup>44</sup> Nelder and Mead<sup>45</sup>, and Law and Bailey<sup>46</sup> are all well known.

In the estimation of parameters for all of the four models, the single rate model, the rate plus non-floating model, the gamma function model and the two-phase model, the whole spectrum of estimation techniques was required.

#### Parameter Estimation for the Single Rate and Rate plus Non-floating Models.

Estimation of the single rate constant of the single rate model and of the non-floating fraction and positive rate constant for the rate plus non-floating model was performed by a derivative method programmed by J.A. Head<sup>47</sup>; the following is a description of the programs mathematical base. The problem is basically the minimization of the sum of squares as defined by

$$\Delta = \sum_{i=1}^N [F(\bar{P}, x_i) - Y_i]^2$$

for  $\Delta$  to be minimized

$$\frac{\partial \Delta}{\partial P_j} = 0 \quad j = 1, 2, 3 \dots m$$

for all 'j' parameters.

Given the initial estimate of  $P_j^0$ ;  $j = 1, 2, 3 \dots m$ , the sum of squares as a function of  $\bar{P}$  may be expanded in an m-dimensional Taylor series about the point  $\bar{P}^0$

$$\Delta(\bar{P}) = \Delta(\bar{P}^0) + \sum_{j=1}^m \left( \frac{\partial \Delta}{\partial P_j} \right) (P_j - P_j^0) + \frac{1}{2} \sum_{j=1}^m \sum_{k=1}^m \left( \frac{\partial^2 \Delta}{\partial P_j \partial P_k} \right) (P_j - P_j^0) (P_k - P_k^0) + \dots$$

Assuming that the initial  $P_j^0$  are near to  $\bar{P}$  which minimize  $\Delta$ , higher order terms may be neglected. For increased accuracy, the equation may be extended but requires increased computation of analytical derivatives.

If we define the following matrices

$$C_j = \frac{\partial \Delta(\bar{P})}{\partial P_j}$$

and

$$A_{jk} = \frac{\partial^2 \Delta(\bar{P})}{\partial P_j \partial P_k}$$

then approximately

$$\Delta(\bar{P}) \approx \Delta(\bar{P}^0) + \sum_{j=1}^m C_j (P_j - P_j^0) + \frac{1}{2} \sum_{j=1}^m \sum_{k=1}^m A_{jk} (P_j - P_j^0) (P_k - P_k^0)$$

The matrices  $C_j$  and  $A_{jk}$  may be computed from the following

$$C_j = \sum_{i=1}^N [F(\bar{P}^0, x_i) - Y_i] \left[ \frac{\partial F(\bar{P}^0, x_i)}{\partial P_j} \right]$$

$$A_{jk} = \sum_{i=1}^N [F(\bar{P}^0, x_i) - Y_i] \left[ \frac{\partial^2 F(\bar{P}^0, x_i)}{\partial P_j \partial P_k} \right] + \left[ \frac{\partial F(\bar{P}^0, x_i)}{\partial P_j} \right] \left[ \frac{\partial F(\bar{P}^0, x_i)}{\partial P_k} \right]$$

from the derivative of  $\Delta$  and assuming  $\frac{\partial \Delta(\bar{P}^0)}{\partial P_m} \approx 0$   
we have

$$\frac{\partial \Delta(\bar{P})}{\partial P_m} = 0 \approx 0 + C_m + \sum_{k=1}^m A_{mk} (P_k - P_k^0)$$

multiplying by the inverse of matrix  $A_{mk}$

$$0 \approx \sum_{k=1}^m A_{km}^{-1} C_m + P_k - P_k^0$$

Through repeated application or iterative procedures, the best estimates of the parameter array may be found.

#### Parameter Estimation for the Gamma Function Model.

The derivative dependent method of Head and the Law and Bailey technique proved unstable in estimation of the batch test kinetic parameters for the gamma function model. An alternative method, that of Powell<sup>48</sup>, proved very successful provided reasonable initial estimates of the parameters were



computed. The initial estimates were computed as follows :

let

$$a = R^2/\sigma^2 - 1$$

and

$$b = R/\sigma^2$$

then the gamma function batch test equation becomes for normalized flows :

$$C = \left( \frac{b}{b+t} \right)^{a+1}$$

taking logarithms of both sides, we have

$$Y = \ln C = (a+1)(\ln b - \ln(b+t))$$

let the function to be minimized be

$$SS = \sum_{i=1}^n (Y_i - (a+1)(\ln b - \ln(b+t_i)))^2$$

at the minimum

$$\frac{dSS}{da} = 0$$

hence

$$\frac{dSS}{da} = 2 \sum_{i=1}^n (Y_i - (a+1)(\ln b - \ln(b+t_i))) (-(\ln b - \ln(b+t_i))) = 0$$

therefore letting

$$X_i = \ln b - \ln(b + t_i)$$

$$\sum_{i=1}^n -Y_i X_i + (a+1) X_i^2 = 0$$

$$a+1 = \frac{\sum_{i=1}^n Y_i X_i}{\sum_{i=1}^n X_i^2}$$

thus a one-parameter search along the 'b' axis will provide a minimum very close to the true minimum; the Powell routine need only start from this near minimum. This initial search weights the smaller of the 'C' or concentration points hence with material which reach nearly 100 percent recovery, the initial minimum is less accurate.

The concentration versus mean residence time curve for a continuously operating bank is described as follows :

$$\underline{\alpha} = \int_0^{\infty} \underline{C}^{-1} \underline{\varepsilon} W(k) dk$$

which may be simplified to

$$\alpha_j = \int_0^{\infty} \frac{W(k) dk}{(1+q_1)^j}$$

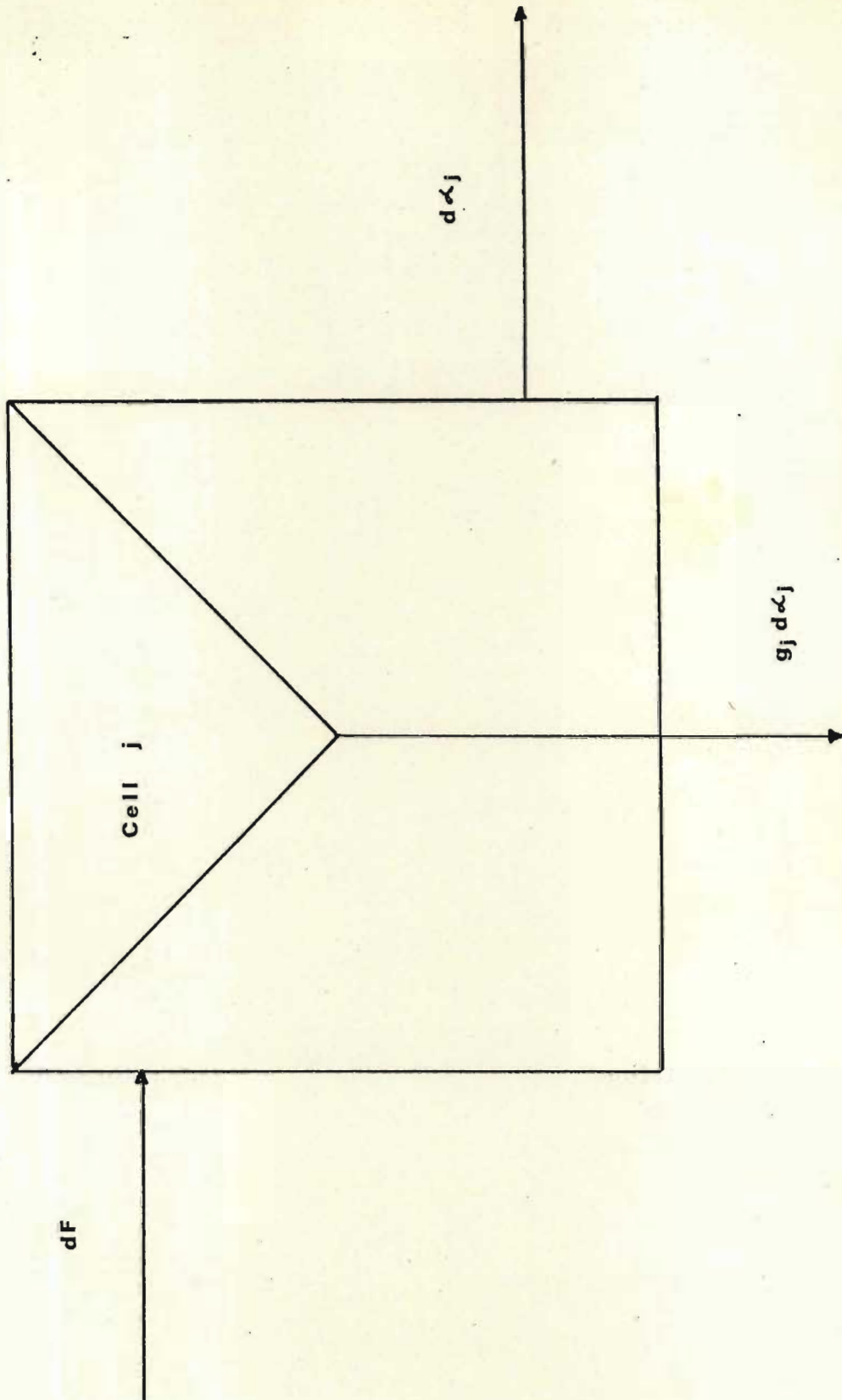
Parameter estimation from continuous test data is lengthy even though the above equation may be rapidly solved by the previously discussed Gauss-Laguerre quadrature technique. The regression was however performed on data from a bank of twenty cells. Initial parameter estimates were obtained by

assuming that the continuous data fit the batch equation, thereafter the non-derivative Powell search method was used to find the minimum sum of squares using the actual continuous operation equation.

#### Parameter Estimation for the Two-Phase Model.

The two-phase model showed poor predictive powers and a large interdependence between parameters. The very flat minimum of the sum of squares surface for batch data required the use of a direct grid search for stable estimation of the two rate parameters. Other slow direct search methods such as that of Hooke and Jeeves<sup>49</sup> would probably prove to be applicable.





53. Flotation Cell

## IX. TEST METHODS AND DATA COLLECTION

In the course of this investigation over fifty batch tests, thirty continuous tests and numerous plant circuit tests were performed. The raw data for batch and continuous tests is available in part two of the thesis. Unless otherwise specifically stated, reagent additions were not changed from normal plant operation. Batch tests may be sub-divided primarily on the basis of their froth removal rates into five groups. In the discussion below the means of data collection and consequent data reduction are described.

All plant batch tests, that is, batch tests performed on plant pulp streams were performed in a five litre Denver cell with a Fagergren impeller. Impeller speed was 2000 rpm. Laboratory tests were performed in a two litre Denver cell with a Denver impeller at 1600 rpm.

All batch and continuous tests are assigned code numbers which aid in the identification and correlation of data.

### Group One - Batch Tests

Included in this group are batch tests on the flotation feed stream in the plant and laboratory tests on crushed and batch ground mill feed.

The general method of weighing and assaying the final products of a batch test was used to generate a recovery versus time curve built back from the final tailings.

Froth removal in these tests was not measured but froth was removed at a rapid and consistent rate.

#### Group Two - Batch Tests

Tests in this group represent a sub-division of batch tests performed on various plant streams at Rustenburg Platinum Mines. The basic problem in using batch test data is how to handle the froth removal. Initially, plant froth removal rates were approximated visually but this method proved to be too indefinite and was abandoned after only a few tests. Thus the froth removal rate was not recorded and was rather arbitrary.

#### Group Three - Batch Tests

Again these batch tests were performed on various plant streams at Rustenburg. These tests represent a slight improvement over the method in group two, in this case the froth removal rate was held constant through the measurement of the wet weight of concentrate removed in a specific time interval and by regulation of the make-up water addition rate. As in group two, the froth was allowed to flow freely over the froth weir and was not mechanically removed by scraping, neither was the froth depth measured.

#### Group Four - Batch Tests

Another improvement was made to the batch test procedure of group three, in this group of plant batch tests, the froth was removed at a measured rate through regulation of the pulp level and use of a froth paddle. The pulp level was measured before and after each concentrate interval. Thus after an approximation of the change in level due to air in the pulp is accounted for, figures for the volume of froth and froth flowrate may be computed for each stage of concen-



trate removal.

Data processing followed the normal batch test procedure with the incorporation of the calculation of froth flowrate/froth volume for application of the two-phase model.

#### Group Five - Batch Tests

In doing the batch tests of the previous groups, it was well understood that the rate or distribution of rates which was assumed to govern flotation of a species, was in fact not being measured; the issue was complicated by the froth phase. In order to closely approximate the rate parameters, the effects of the froth phase must be minimized by very rapid froth removal. Batch tests of this group were performed with this in mind. The froth was removed by a hand froth paddle as rapidly as was physically possible. In most instances pulp height and the wet weight of concentrate were recorded for each concentrate interval.

Data processing followed the normal procedure for batch testing with the calculation of froth flowrate/froth volume needed for application of the two-phase model.

#### Group Six - Continuous Tests

A number of tests were conducted on a continuous flotation bank, in order to test the various models' abilities in the continuous case.

The production of residence time recovery curves from a bank of continuously operating cells is statistically difficult, particularly with a low-grade ore. Several methods may be used to obtain data for a recovery curve for a bank of cells; two methods were attempted at Rustenburg. The

simplest method of obtaining a recovery curve is to take a pulp sample from each cell composed of small samples from various points below the froth layer and assay it for the particular species. By ignoring concentrate tonnages (they are usually small compared to tailings tonnages), the recovery curve may be produced directly. Unfortunately, samples taken in this manner are extremely scattered even when considerable care is taken in the sampling procedure. (Figure 54)

The above method was abandoned in favour of a slightly better method. This second method consists of timing samples of concentrate from each cell, then weighing and assaying them. Measurements from a magnetic flowmeter and a gamma density gauge provide a feed tonnage and several samples of the tailings stream are taken for assay. The recovery-residence time curve is calculated directly from the feed tonnage, concentrate tonnages, concentrate assays and the mean tailings assay. Due to the cell and launder design, it is impossible to collect the entire concentrate flow for a specified time thus a correction factor must be incorporated.

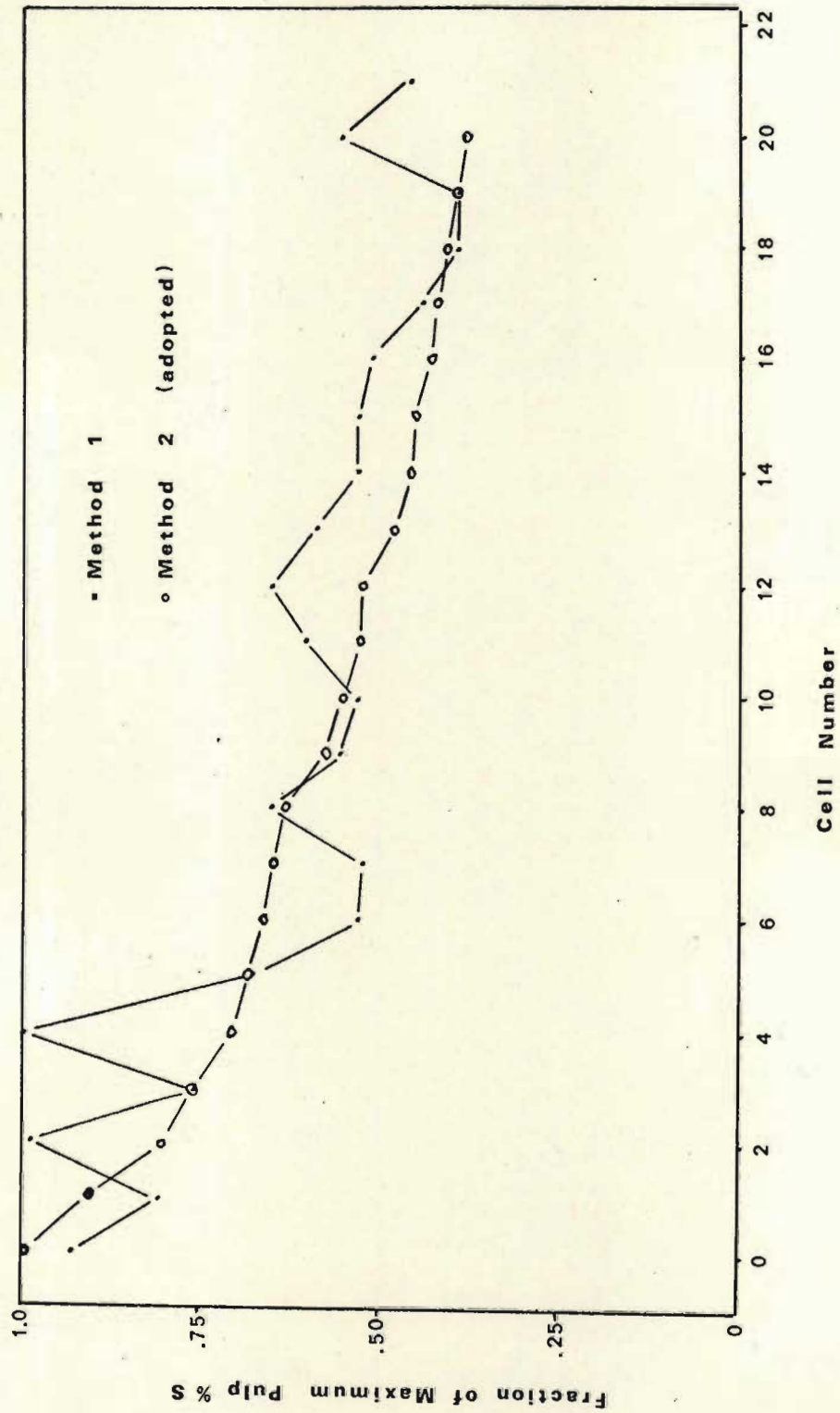
#### Group Seven - Special Tests

This group contains batch tests at various froth removal rates and are used determining the effect of froth removal upon rate parameters of the single phase models. Froth depth was not measured but a set wet weight of concentrate was removed in each time interval.

#### Data for Various Flotation Circuits

This eighth data group is not included in part two of the thesis, the data is used as a check on the accuracy of circuit simulations by the gamma function model. Data consists of assays and measured flows (density and volume) of all main flotation streams for any specified flotation circuit. Samples for assay from each stream were hand-cut every fifteen or twenty minutes for an eight or sixteen hour test. Every effort was made to ensure steady operation of the entire plant and consistently accurate sampling. No attempts were made at determining the response of the flotation circuit to any variables as steady-state operation was the primary objective. Thus the flotation data represents an accurate analysis of the flotation circuit kinetics for a long steady-state period.





54. Continuous Operation Curves

LIST OF SYMBOLS

- $a_{ij}$  - fraction of the tailings material from cell  $j$  which is fed to cell  $i$ .
- $b_{ij}$  - fraction of the concentrate material from cell  $j$  which is fed to cell  $i$ .
- $C$  - concentration of a species.
- $C_0$  - initial concentration.
- $C_n$  - non-floatable fraction of a species.
- $\underline{C}$  - configuration matrix.
- $E(t)$  - residence time distribution.
- $\underline{F}$  - column matrix of species distribution in a flotation feed stream.
- $g_j$  - ratio of concentrate material to tailings material for cell  $j$ .
- $k$  - first-order rate constant. ( $\text{min}^{-1}$ )
- $\bar{k}$  - mean of the modified gamma function distribution of first-order rate constants. ( $\text{min}^{-1}$ )
- $n$  - number of flotation cells or perfectly stirred units.
- $\underline{r}$  - a column matrix which contains the partial fractional recovery from each flotation unit.
- $R$  - total fractional recovery of a species.

LIST OF SYMBOLS (Continued)

$t$	-	time
$T$	-	mean retention time in a continuous cell.
$v$	-	volumetric flowrate to a set of cells.
$V$	-	effective volume of a single stirred unit.
$W(k)$	-	the distribution of first-order rate constants in the gamma function model.
$W_i$	-	weights in the Gauss-Laguerre integration formula.
$j$	-	tailings material flow from a cell $j$ .
$\varepsilon$	-	column matrix of fractional feed flows to various flotation cells.
$\Gamma$	-	gamma function.
$\sigma^2$	-	variance of the gamma function model distribution of rate constants. ( $\text{min}^{-2}$ )
$\mu$	-	microns.
$\beta$	-	power of $X$ in the general integration form.
$X_i$	-	roots in the integration solution.



XI. BIBLIOGRAPHY

- 1) Truter, F.C. Modern Concepts of the Bushveldt Igneous Complex. CCTA Southern Reg. Comm. Geol., Salisbury 1, 77-92.
- 2) Wager L.R. and Brown, G.M. Layered Igneous Rocks. Oliver and Boyd, London 1968.
- 3) Hamiliton, Warren. Bushveldt Complex - Product of Impacts? The Geological Society of South Africa, Special Publication No. one, South Africa, 1970. Symposiums on the Bushveldt Igneous Complex and Other Layered Intrusions, July, 7-14, 1969. Ed. D.J.L. Visser and G. Von Gruenewaldt.
- 4) Du Toit, Alex L. The Geology of South Africa. Oliver and Boyd, Edinburgh, 1954.
- 5) Schmidt, E.R. The Structure and Composition of the Merensky reef on the Rustenburg Platinum Mine, Pretoria University, 1948.
- 6) Johannesburg Consolidated Investments Minerals Processing Laboratory. Project 4.1A No. 2.
- 7) Zuniga, H.G. The efficiency obtained by flotation is an exponential function of time. Bol. miners Soc. nacl. Minería (Chile) Vol. 47, 1935, pp. 83 - 86.
- 8) Schuhmann, R. Flotation Kinetics. J. Methods for steady-state study of flotation problems. J. Phys. Chem. Vol. 46, 1942, pp. 891 - 902.

XI. BIBLIOGRAPHY (Continued)

- 9) Tomlinson, H.S. and Fleming, M.G. Flotation rate studies. International Mineral Processing Congress, 6th Cannes 1963. Oxford. Pergamon, 1965, pp. 563-579.
- 10) Sutherland, K.L. Physical Chemistry of Flotation Part XI. Kinetics of the Flotation Process. J. Phys. Colloid Chem. vol. 52, 1948, pp. 394-425.
- 11) Evans, L.F. Bubble-mineral attachment in Flotation. Indus. Engineering Chem. vol. 46, No. 11, 1964. pp. 2420-2429.
- 12) Egeles, M.A. Theoretical basis of Flotation of non-sulphide Minerals. Metallurg. 1950.
- 13) Bushell, C.H.G. Kinetics of Flotation. Trans. Soc. Min. Engrs. (AIME) vol. 223, 1962, pp. 266-278.
- 14) Arbiter, N. and Harris, C.C. Flotation Kinetics. Fuerstenau D.W. ed. Froth Flotation 50th Anniversary Vol. New York, AIME. 1962. Chapt. 8, pp. 215-262.
- 15) c.f. 7
- 16) Imaizumi, T. and Inoue, T. Kinetic Considerations of froth flotation. International Mineral Processing Congress. 6th Cannes, 1963. Ed. A. Roberts, Oxford Pergamon, 1965. pp. 581-593.
- 17) Loveday, B.K. Analysis of Froth Flotation Kinetics. Trans. Inst. Min. Metall. vol. 75, 1966, pp. C219-C225.

XI. BIBLIOGRAPHY (Continued)

- 18) King, R.P. An Improved Distributed-Parameter Model for the Kinetics of Mineral Flotation. National Institute of Metallurgy, Republic of South Africa, Report No. 996, August 1970.
- 19) Woodburn, E.T. Mathematical Modelling of Flotation Processes. Minerals Science and Engineering. April, 1970. pp. 3 - 17.
- 20) Wark, I.W. Principles of Flotation. Australasian Institute of Mining and Metallurgy 1938. pp. 44.
- 21) c.f. 8
- 22) c.f. 14
- 23) Maksimov, I.I. and Khainmin, V. YA. Effect of Processes occurring in the froth layer on the rate and selectivity of Flotation. Soviet J. Non-ferrous metals, vol. 6, No. 5, 1965, pp. 8-10.
- 24) c.f. 14
- 25) Taggart, A.F. Handbook of Mineral Dressing. John Wiley and Sons Inc. New York, 1954. pp. 12-29.
- 26) c.f. 20 pp. 177.
- 27) Jowett, A. Investigation of Residence Time of fluid in Froth Flotation Cells. Brit. Chem. Engineering, vol. 6, 1961, pp. 254-258.



XI. BIBLIOGRAPHY (Continued)

- 28) Woodburn, E.T., King, R.P. and Colborn, R.P. The effect of Particle Size Distribution on the Performance of a Phosphate Flotation Process. Metallurgical Trans. vol. 2, November 1971, pp. 3163 - 3174.
- 29) c.f. 17
- 30) Smith, S.W., Short, R.C. and Snyman, G.C. Residence-Time Measurements on Flotation Cells using Radioactive Platinum Ore. Atomic Energy Board of South Africa. Report No. PIN 237 (BR) Restricted. August, 1974.
- 31) c.f. 16
- 32) Black, K.G. and Faulkner, B.P. Evaluation of Batch Flotation Results by Multiple Linear Regression. AIME Trans. March 1972, vol. 252 pp. 19,20.
- 33) Harris, C.C. and Chakravarti, A. Semi-Batch Froth Flotation Kinetics: Species Distribution Analysis. SME Transactions, vol. 247, June 1970. pp. 162-172.
- 34) Taggart, A.F. pp. 12 - 53.
- 35) The Determination of Total Phenols (Tar Acids). Colorimetric Chemical Analytical Methods. pp. 65 - 66.
- 36) c.f. 14
- 37) Harris, C.C. and Rimmer, H.W. Study of a two-phase Model of the Flotation Process. Trans. Inst. Min. Metall., vol. 75, 1966. pp. C153 - C162.

XI. BIBLIOGRAPHY (Continued)

- 38) Woodburn, E.T. and Loveday, B.K. Effect of Variable Residence Time Data on the performance of a Flotation System. J.S. Afr. Inst. Min. Metall., 1965. pp. 612 - 628.
- 39) Dieft, P.A. The Mathematics Associated with Flotation Networks. University of Natal, South Africa, July, 1970.
- 40) c.f. 37
- 41) Fletcher, R. and Powell, M.J.D. A rapidly convergent descent method for Minimization. Computer J. Vol. 6, 1963-64. pp. 163 - 168.
- 42) Powell, M.J.D. An efficient method for finding the Minimum of a Function of Several Variables without calculating Derivatives. Computer J. Vol. 7, 1964-65. pp. 155 - 162.
- 43) Rosenbrock, H.H. An automatic method for finding the greatest or least Value of a Function. Computer J. Vol. 3, 1960. pp. 175 - 184.
- 44) Hooke, R. and Jeeves, T.A. "Direct Search" Solution for Numerical and Statistical Problems. Association for Computing Machinery. Vol. 8, 1961. pp. 212-229.
- 45) Nelder, J.A. and Mead, R. A simplex method for Function Minimization. Computer Journal, Vol. 7, 1964-65. pp. 308 - 313.
- 46) Law and Bailey. Chem. Eng. Science. Vol. 18 (1963) pp. 189 - 202.

XI. BIBLIOGRAPHY (Continued)

- 47) Head, J.A. Fitit, a computer program to least squares fit non-linear theories. National Technical Information Service, U.S. Dept. of Commerce, Dec. 1970.
- 48) c.f. 42
- 49) c.f. 44
- 50) Gaudin, A.M. Flotation. McGraw-Hill, New York, 1957. pp. 363.



INDUSTRIAL APPLICATION  
OF A  
FLOTATION MODEL

PART TWO

## TABLE OF CONTENTS

I.	RAW TEST DATA	
A.	GROUP I BATCH TESTS .....	PAGE 168
B.	GROUP II BATCH TESTS .....	171
C.	GROUP III BATCH TESTS .....	174
D.	GROUP IV BATCH TESTS .....	177
E.	GROUP V BATCH TESTS	
	1. ROUGHER TO ROUGHER CONCENTRATE PREDICTIONS .....	182
	2. BATCH TO CONTINUOUS PREDICTIONS .....	187
F.	GROUP VI CONTINUOUS TESTS	
	1. GENERAL .....	190
	2. BATCH TO CONTINUOUS PREDICTIONS .....	197
G.	GROUP VII SPECIAL TESTS	
	1. FROTH REMOVAL DATA .....	201
II.	REGRESSION PROGRAMS	
A.	MAINLINE PROGRAMS	
	1. FTSRT/BLOK1/SETUP/ANYFT .....	203
	A. S/R THEO AND DERIV .....	210
	B. DATA FORMAT .....	214
	2. REGAL .....	215
	A. MODSQ (RATE PLUS NONFLOATING MODEL) .....	216
	B. DATA FORMAT .....	217
	3. REGDL .....	218
	A. MODSQ (GAMMA FUNCTION MODEL) .....	221
	B. DATA FORMAT .....	222
	4. INT/SEAR2 .....	223
	A. DATA FORMAT .....	226
	5. GCQNR .....	227
	A. MODSQ (CONTINUOUS GAMMA FORM) .....	229
	B. DATA FORMAT .....	230
B.	GROUP SPECIFIC SUBROUTINES	
	1. GRIDS .....	231
	2. VA02A .....	232
	3. VA01A .....	242
	4. AMIN1 .....	245
	5. DIST .....	246
	6. DIST1 .....	247



## III. PREDICTIVE PROGRAMS

## A. MAINLINE PROGRAMS

1. BCONT .....	248
A. DATA FORMAT .....	249
2. GCONT .....	250
A. DATA FORMAT .....	252
3. CLEAN .....	253
A. DATA FORMAT .....	256
4. ROCLF .....	257
A. DATA FORMAT .....	258
5. ITIT .....	259

## IV. CIRCUIT ANALYSIS PROGRAMS

## A. MAINLINE PROGRAMS

1. PRFW0/PRFW1/PRFW2 .....	260
A. IAIB (CIRCUIT 36) .....	264
B. IAIB (CIRCUIT 48) .....	265
C. IAIB (CIRCUIT 53) .....	266
D. IAIB (CIRCUIT 61) .....	267
E. DATA FORMAT .....	268
2. PRF22/PRF23 .....	269
A. IAIB (VARIOUS CIRCUITS) .....	273

## V. GENERAL SERVICE PROGRAMS

## A. MAINLINE

1. RESTT .....	276
2. USEIN .....	277
3. BATPL .....	278
4. PYRHH .....	280

## B. GROUP SPECIFIC SUBROUTINES

1. QTFG .....	282
2. SSQS .....	283

## VI. GENERAL USAGE SUBROUTINES

A. BREAD .....	284
B. MINV5 .....	285
C. RCGAM .....	286
D. KROPH/LAGUR/LROOT/LRECR .....	287
E. RFACT .....	292
F. MODSQ (TWO PHASE MODEL) .....	293





ALL OF THE FOLLOWING DATA IS IN THE FORM OF NORMALIZED CONCENTRATION CURVES VERSUS TIME (MEAN TIME FOR CONTINUOUS OPERATION)  
 FOR EACH TEST THERE EXISTS A HEADER CARD WHICH ROUGHLY IDENTIFIES THE TEST FOLLOWED BY A DATA CARD (2) - THE THIRD CARD (AND FOURTH WHERE NDATA IS GREATER THAN 20) CONTAINS THE NORMALIZED CONCENTRATIONS IN A FIELD OF FOUR, I.E. .526 IS WRITTEN AS 0526 - THE FOURTH CARD (OR FIFTH AND SIXTH) CONTAINS THE CORRESPONDING TIMES ( 0500 IS 5.0 MINS )

THE IDENTIFIER CARD (ONE) CONTAINS A SERIES OF IDENTIFYING NUMBERS AS  
 HEADER CARD - ALPHA INFORMATION ABCDDEEEFEGGG

A IDENTIFYS THE PLANT STREAM TESTED OR ITS EQUIVALENT

1 EQUALS FLOTATION FEED \*\*\* 2 EQUALS FLOTATION FINAL TAILING  
 3 EQUALS ROUGHER CONCENTRATE \*\*\* 4 EQUALS CLEANER FEED  
 5 EQUALS CLEANER MIDDINGS RETURN \*\*\* 6 EQUALS CLEANER TAILINGS  
 7 IS NON EXISTENT \*\*\* 8 EQUALS ROUGHER FEED ( 1 PLUS 6)

B INDICATES THE SPECIES ASSAYED FOR

1 EQUALS TOTAL SULFUR VALUE \*\*\* 2 EQUALS NONSULFIDE MATERIAL  
 3 EQUALS CHALCOPYRITE (CU) \*\*\* 4 EQUALS PENTLANDITE (NI)  
 5 EQUALS THE CALCULATED PYRRHOTITE (FE7S8)

C INDICATES THE SIZE FRACTION FOR THAT SPECIES

1 EQUALS ALL SIZES TOGETHER \*\*\* 2 EQUALS THE RANGE GREATER 208 MICRONS  
 3 EQUALS THE RANGE 208 TO 104 MICRONS \*\*\* 4 EQUALS 104 TO 53 MICRONS  
 5 EQUALS MATERIAL LESS THAN 53 MICRONS

DD IS THE TEST RUN NUMBER IF ANY

EEEE IS THE DATE OF THE TEST - DAY/MONTH

F SIGNIFYS THE TEST MODE \*\*\* 1= PLANT BATCH 2=CONTINUOUS BANK  
 3= LABORATORY BATCH

GGGG IS THE TEST TIME IN HOURS AND MINUTES



## \*\*\*\*\* GROUP ONE \*\*\*\*\*

THE FOLLOWING TEST IS USED IN CIRCUIT ANALYSIS \* SEE PAGES 115-127

FL0T FEED BATCH 2303 NO 1 SULF	111	23031
011		
10000782066005370438038103340294026402370215		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 1 NON SULF	121	23031
011		
10000992098609820976097309700966096209590956		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 1 COPPER	131	23031
011		
10000505034902760235021301970183017101610153		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 1 NON CU	161	23031
011		
10000991098509800974097109670963095909550952		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 1 NICKEL	141	23031
011		
10000892073105770474042403870358034103270317		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 1 NON NI	171	23031
011		
10000991098509800975097109670963096009560953		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH NO 1 FES2	151	23031
011		
10000834073606060491042003610309026902310202		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 2 SULFUR	111	23031
011		
10000629046403780322028702630238022002080199		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 2 NON SULF	121	23031
011		
10000988098009730967096209580953094909460942		
00000500100015002000250030003500400050006000		



FL0T FEED BATCH 2303 NO 2 COPPER	131	23031
011		
10000358026502290206019101800169016001520145		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 2 NON CU	161	23031
011		
10000987097809700964095909550950094609430939		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 2 NICKEL	141	23031
011		
10000692046103920360034503360328032103160311		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH 2303 NO 2 NON NI	171	23031
011		
10000987097809710965096009560951094709430940		
00000500100015002000250030003500400050006000		
FL0T FEED BATCH NO 2 FES2	151	23031
011		
10000714054604320354030402700232020701910179		
00000500100015002000250030003500400050006000		

THE FOLLOWING TESTS ARE USED IN A COMPARISON OF LABORATORY BATCH FLOTATION  
AND PLANT BATCH FLOTATION \* SEE PAGES 25-36

MINERAL LAB TEST SULFUR 3409/1	111	33409
04013		
1000025701750145012901160109010400980093009000880084		
0000050010001500200025003000350040004500500055006		
MINERAL LAB TEST SULFUR 3409/2	111	33409
04013		
1000030601960157013401190112010600920085008100780076		
0000050010001500200025003000350040004500500055006		
MINERAL LAB TEST COPPER 3409/1	131 1	33409
04013		
1000020201690155014601380134013001270124012201200116		
0000050010001500200025003000350040004500500055006		
MINERAL LAB TEST COPPER 3409/2	131 2	33409
04013		
1000020401640148013701300126012201150111010801060104		
0000050010001500200025003000350040004500500055006		
MINERAL LAB TEST NICKEL 3409/1	141 1	33409



04013		
1000038503380320031103050301029802950292029102890286		
0000050010001500200025003000350040004500500055006		
MINERAL LAB TEST NICKEL 3409/2	141 2	33409
04013		
1000042703600339032603180313030903010297029402920290		
0000050010001500200025003000350040004500500055006		
MINERAL LAB TEST FE7S8 3409/1	151 1	33409
04013		
100002330120008100590042003200250018001100080004		
0000050010001500200025003000350040004500500055006		
MINERAL LAB TEST FE7S8 3409/2	151 2	33409
04013		
100003050157010500740055004500380020001100060003		
0000050010001500200025003000350040004500500055006		
MINERAL BATCH LAB 100/NORITE 3733	111 1	33733
04013		
1000028202420225021502050198019301870183017901760172		
0000050010001500200025003000350040004500500055006000		
MINERAL BATCH LAB 100/NORITE 3733	111 2	33733
04013		
1000032102750247023202210213020601990194019001850181		
0000050010001500200025003000350040004500500055006		
MINERAL BATCH LAB 100/REEF 3737	111 1	33737
04013		
1000047503900345032103060293028402760268026102530246		
0000050010001500200025003000350040004500500055006		
MINERAL BATCH LAB 100/REEF 3737	111 2	33737
04013		
1000056304630407036703460326031303030293028602770269		
0000050010001500200025003000350040004500500055006		
MINERAL BATCH LAB 100/MERENSKY 3742	111 1	33742
04013		
1000041803020249021301890172016001470138012901230115		
0000050010001500200025003000350040004500500055006		
MINERAL BATCH LAB 100/MERENSKY 3742	111 2	33742
04013		
1000042802960220018201610145013301230114010601000094		
0000050010001500200025003000350040004500500055006		



## \*\*\*\*\* GROUP TWO \*\*\*\*\*

THE MAJORITY OF THESE TESTS ARE USED TO ILLUSTRATE THE RATES OF FLOTATION  
IN VARIOUS PLANT FLOTATION STREAMS \* SEE PAGES 44-46

MINERAL BATCH RD CONC 46	31146210911711
04007100800500900010	
1000052303840299025302130151	
0000050010001500200025003500	
GANGUE BATCH RD CONC 46	32146210911711
04007102001103001010	
1000086707920735069806600597	
0000050010001500200025003500	
MINERAL BATCH CL FEED 46	41146210911515
04007100800500900010	
1000048903660303024401890133	
0000050010001500200030004000	
GANGUE BATCH CL FEED 46	42146210911515
04007102001103001010	
1000081207340682063305760521	
0000050010001500200030004000	
MINERAL BATCH CL MID 44	51144070911850
04006102001103000010	
100005660374013200630034	
0000050010002000300040005000	
MINERAL BATCH CL MID 46	51146210911625
04007102001103001010	
1000074305720478038503450233	
0000050010001500200025003500	
GANGUE BATCH CL MID 44	52144070911850
04006102001103000010	
100007520579033902260167	
0000050010002000300040005000	
GANGUE BATCH CL MID 46	52146210911625
04007102001103001010	
1000087807760710063205970507	
0000050010001500200025003500	
MINERAL BATCH CL TAILS 44	61144070911610



04006102001103000010	
100007710671055804430332	
000005001300200030004000	
MINERAL BATCH CL TAILS 46	61146210912019
04007100800500900010	
1000084008030724064506040487	
0000050010001500200030004000	
GANGUE BATCH CL TAILS 44	62144070911610
04006102001103000010	
100008880813077707300692	
000005001300200030004000	
GANGUE BATCH CL TAILS 46	62146210912019
04007102001103001010	
1000093109180887086108460807	
0000050010001500200030004000	
MINERAL BATCH RD FEED 44	81144070911700
04007100800500900010	
1000071506280538050004590422	
0000050010002000300040005000	
MINERAL BATCH RD FEED 46	81146210911912
04007100800500900010	
1000069705600530048904560415	
0000050010002000300040005000	
GANGUE BATCH RD FEED 44	82144070911700
04007102001103001010	
1000098209710953094509380929	
0000050010002000300040005000	
GANGUE BATCH RD FEED 46	82146210911912
04007102001103001010	
1000098509760973096709620956	
0000050010002000300040005000	
MINERAL BATCH RD FEED WET -208	81 49131011102
04007	
1000080206490591046703010251	
000005001000150020003000400050006000	
MINERAL BATCH RD FEED WET +104	81 491310111453
04007	
1000080607190653055604950458	
000005001000150020003000400050006000	
GANGUE BATCH RD FEED WET -208	82 49131011102

04007

1000098309700960094709290912

000005001000150020003000400050006000

GANGUE BATCH RD FEED WET +104

82 49131011453

04007

1000098309700953094009220908

000005001000150020003000400050006000



## \*\*\*\*\* GROUP THREE \*\*\*\*\*

FROTH WET WEIGHTS WERE CONTROLLED THROUGH PULP LEVEL AND MEASURED  
 I.E. 'PULL 100' MEANS 100 GRAMS OF WET CONCENTRATE COLLECTED IN 5 MINUTES  
 A FROTH PADDLE WAS NOT USED

GANGUE BATCH RD FEED +208	PULL 200/5	82200041111600
040071		
1000099809920990098709720951		
0000050010001500200030004000		
GANGUE BATCH RD FEED +104	PULL 200/5	82300041111600
040071		
1000099709940993099109880985		
0000050010001500200030004000		
GANGUE BATCH RD FEED +53	PULL 200/5	82400041111600
040071		
1000099509930992099009880985		
0000050010001500200030004000		
GANGUE BATCH RD FEED -53	PULL 200/5	82500041111600
040071		
1000097609630950093909200907		
0000050010001500200030004000		
MINERAL BATCH RD FEED +208	PULL 200/5	81200041111600
040071		
1000095108720834079506010425		
0000050010001500200030004000		
MINERAL BATCH RD FEED +104	PULL 200/5	81300041111600
040071		
1000079507520731069706500617		
0000050010001500200030004000		
MINERAL BATCH RD FEED +53	PULL 200/5	81400041111600
040071		
1000076107170693067006320607		
0000050010001500200030004000		
MINERAL BATCH RD FEED -53	PULL 200/5	81500041111600
040071		
1000070606220539047303960350		
0000050010001500200030004000		
MINERAL BATCH RD CONC PULL 150		31100061110300
040071		



1000064104760366028502400194	
0000050010001500200025003500	
MINERAL BATCH RD CONC PULL 100	31100061110525
040071	
1000067704690345024701950154	
0000050010001500200025003500	
GANGUE BATCH RD CONC PULL 150	32100061110300
040071	
1000096809480926090008790853	
0000050010001500200025003500	
GANGUE BATCH RD CONC PULL 100	32100061110525
040071	
1000096709420924090408890869	
0000050010001500200025003500	
MINERAL BATCH RD FEED 47 PULL 180	81147280911210
04007101000501401010	
1000075406440589048103600293	
0000050010001500200030004000	
MINERAL BATCH RD FEED 47 PULL 350	81147280911415
04007100900501401010	
1000069605990545044403420292	
0000050010001500200030004000	
MINERAL BATCH RD FEED 47 PULL 100	81147280911602
04007101000501401010	
1000077006960643053804890423	
0000050010001500200030004000	
MINERAL BATCH RD FEED 47 PULL 220	81147280911650
04007101000501401010	
1000066406330557042503470275	
0000050010001500200030004000	
MINERAL BATCH RD FEED 48 PULL 100	81148051011206
04011101000501401010	
10000810075407190633057705410523050004830467	
00000500100015002000300040005000600070008000	
MINERAL BATCH RD FEED 48 PULL 300	81148051011340
04010101000501401010	
1000063305660538046304250404037003460324	
0000050010001500200030004000500060007000	
MINERAL BATCH +208 WET RD FEED 48 PULL 180	81248051011544
04007101000501401010	

1000074606990677060405620536	
0000050010001500200030004000	
GANGUE BATCH RD FEED 47 PULL 180	82147280911210
04007101000501401010	
1000099109820976096709560942	
0000050010001500200030004000	
GANGUE BATCH RD FEED 47 PULL 350	82147280911415
04007100900501401010	
1000098109710963095409390929	
0000050010001500200030004000	
GANGUE BATCH RD FEED 47 PULL 100	82147280911602
04007101000501401010	
1000099309890984097709720965	
0000050010001500200030004000	
GANGUE BATCH RD FEED 47 PULL 220	82147280911650
04007101000501401010	
1000098609830975095909470937	
0000050010001500200030004000	
GANGUE BATCH RD FEED 48 PULL 100	82148051011206
04011101000501401010	
10000995099209900986098109780975097309700968	
00000500100015002000300040005000600070008000	
GANGUE BATCH RD FEED 48 PULL 300	82148051011340
04010101000501401010	
1000098209730964095509440933092009120908	
0000050010001500200030004000500060007000	
GANGUE BATCH +208 WET RD FEED 48 PULL 180	82248051011544
04007101000501401010	
1000099009860984097709680961	
0000050010001500200030004000	



## \*\*\*\*\* GROUP FOUR \*\*\*\*\*

THESE TESTS ARE USED PRIMARILY IN CONNECTION WITH THE TWO PHASE FLOTATION  
 MODEL \* SEE PAGES 88-103  
 FROTH HEIGHT AND FROTH FLOW WERE MEASURED AND CONTROLLED

MINERAL BATCH RD CONC 100/5 NUTST	31100020212130
04007	
1000078205790382023801280079	
0000050010001500200030004000	
MINERAL BATCH RD CONC 100/5 NWTST	31100070212110
04007	
1000071705300375025001450096	
0000050010001500200030004000	
MINERAL BATCH RD CONC 100/5 NWTST	31100110212145
04007	
1000077505760419031901960111	
0000050010001500200030004000	
MINERAL BATCH RD CONC 100/5 NWTST	311 120312110
04005	
1000074206270478036802440177	
00000500100015002000300040005000	
MINERAL BATCH RD CONC 100/5 NWTST	31100130212035
04007	
1000061204040294021501770100	
0000050010001500200030004000	
MINERAL RD CONC 100/5 NWTST	311 220212155
04005	
1000075605800496039402370162	
00000500100015002000300040005000	
RD CONC MINERAL 100/5	311 260212155
007	
1000075605800496039402370162	
0000050010001500200030004000	
GANGUE BATCH RD FEED 100/5 NWTST	32100020212130
04007	
1000093708790819077307140674	
0000050010001500200030004000	
GANGUE BATCH RD CONC 100/5 NWTST	32100070212110



04007	
1000096009160872081407750738	
000005001000150020003000400050006000	
GANGUE BATCH RD CONC 100/5 NWTST	32100110212145
04007	
1000094708930831077606860617	
0000050010001500200030004000	
GANGUE BATCH RD CONC 100/5 NWTST	321 120312110
04005	
1000095709290891085207880743	
00000500100015002000300040005000	
GANGUE BATCH RD CONC 100/5 NWTST	32100130212035
04007	
1000096009240896086808390799	
0000050010001500200030004000	
GANGUE RD CONC 100/5 NWTST	321 220212155
04005	
1000096209290899087208110770	
00000500100015002000300040005000	
RD CONC GANGUE 100/5	321 260212155
007	
1000096209290899087208110770	
0000050010001500200030004000	
MINERAL BATCH RD FEED 200/5	81100010212025
04007	
1000082007140655057505290447	
0000050010001500200030004000	
MINERAL BATCH RD FEED 100/5 NWTST	81100020212030
04007	
1000074406620614053504760436	
0000050010001500200030004000	
MINERAL BATCH RD FEED 100/5 NWTST	81100070212000
04007	
10000685059605370504044303890	
0000050010001500200030004000	
MINERAL BATCH RD FEED 100/5 NWTST	81100110212040
04007	
1000065205750435039303390303	
0000050010001500200030004000	
MINERAL BATCH RD FEED 100/5 NWTST	811 120311905
04007	

1000081507070646061505570503	
00000500100015002000300040005000	
MINERAL BATCH RD FEED 100/5 NWTST	81100130211915
04007	
1000076206320576053804770431	
0000050010001500200030004000	
MINERAL BATCH RD FEED 100/5 NWTST	811 260212030
04007	
1000073206570551050404360407	
00000500100015002000300040005000	
GANGUE BATCH RD FEED 200/5	82100010212025
04007	
1000098809790971096309530942	
0000050010001500200030004000	
GANGUE BATCH RD FEED 100/5 NUTST	82100020212030
04007	
1000099309870982097609660959	
0000050010001500200030004000	
GANGUE BATCH RD FEED 100/5 NWTST	82100070212000
04007	
1000099009840977097209630956	
0000050010001500200030004000	
GANGUE BATCH RD FEED 100/5 NWTST	82100110212040
04007	
1000098809800971096509560948	
0000050010001500200030004000	
GANGUE BATCH RD FEED 100/5 NWTST	821 120311905
04007	
1000099509910987098409790974	
00000500100015002000300040005000	
GANGUE BATCH RD FEED 100/5 NWTST	82100130211915
04007	
1000099309880982097809700963	
0000050010001500200030004000	
GANGUE BATCH RD FEED 100/5 NWTST	821 260212030
04007	
1000099209860981097609690964	
00000500100015002000300040005000	
GANGUE BATCH RD CONC INFINITE NWTST	321 010311800
04005	



10000839077307410718	
00000500100015002000300040005000	
GANGUE BATCH RD CONC INFINITE NWTST	321 080312055
04005	
10000811069906310600	
00000500100015002000300040005000	
GANGUE BATCH RD CONC INFINITE NWTST	32100140211930
04005	
10000805070306500617	
00000500100015002000	
GANGUE BATCH RD CONC INFINITE NWTST	321 200211945
04005	
10000751063505570520	
00000500100015002000300040005000	
GANGUE BATCH RD CONC INFINITE NWTST	321 220212045
04005	
10000752064705980566	
00000500100015002000300040005000	
GANGUE BATCH RD FEED INFINITE NWTST	821 010311625
04007	
1000096709540932092109050893	
00000500100015002000300040005000	
GANGUE BATCH RD FEED INFINITE NWTST	821 080311935
04007	
1000096609520934092309070898	
00000500100015002000300040005000	
GANGUE BATCH RD FEED INFINITE NWTST	82100140211805
04007	
1000097009500927091408950882	
0000050010001500200030004000	
GANGUE BATCH RD FEED INFINITE NWTST	821 200211810
04007	
10000955094409140894087308590851	
00000500100015002000300040005000	
GANGUE BATCH RD FEED INFINITE NWTST	821 220211920
04007	
1000096809420919090508840873	
00000500100015002000300040005000	



C THE FOLLOWING SIZE CURVES WERE BUILT FROM COMPOSITES OF THE SAMPLES FROM  
 C TESTS 321--010311800 321--080312055 321--140211930 321--200211945 AND  
 C 321--220212045

C  
 GANGUE BATCH RC INF +208 322 12400  
 04005  
 10000962095709560955  
 00000500100015002000300040005000  
 GANGUE BATCH RC INF +104 323 12400  
 04005  
 10000911089008790879  
 00000500100015002000300040005000  
 GANGUE BATCH RC INF +53 324 12400  
 04005  
 10000892085108350829  
 00000500100015002000300040005000  
 GANGUE BATCH RC INF -53 325 12400  
 04005  
 10000775065905940556  
 00000500100015002000300040005000

C  
 C THE FOLLOWING SIZE CURVES WERE BUILT FROM COMPOSITES OF THE SAMPLES FROM  
 C TESTS 821--010311935 821--080311935 821--140211805 821--200211810 AND  
 C 821--220211920

C  
 GANGUE BATCH INFINITE +208 822 12400  
 04007  
 1000099809980996099609960996  
 00000500100015002000300040005000  
 GANGUE BATCH INFINITE +104 823 12400  
 04007  
 1000099609950992099109910991  
 00000500100015002000300040005000  
 GANGUE BATCH INFINITE +53 RF 824 12400  
 04007  
 1000099309920989098709850985  
 00000500100015002000300040005000  
 GANGUE BATCH RF INF -53 825 12400  
 04007  
 1000092108810831080007580731  
 00000500100015002000300040005000

## \*\*\*\*\* GROUP FIVE \*\*\*\*\*

THESE TESTS THAT FOLLOW ARE USED IN ASSESSING THE VALIDITY OF ' K  
CONSISTENCY ' \* SEE PAGES 54 AND 71

BATCH RD FEED COPPER 8/8/74 NO DEXTRIN TO RC	831	080811620
04007		
1000040603360283025402200203		
0000050010001500200030004000		
BATCH RD FEED NICKEL 8/8/74 NO DEXTRIN TO RC	841	080811620
04007		
1000066304850390035803140298		
0000050010001500200030004000		
BATCH RD FEED SULF 8/8/74 NO DEXTRIN TO RC	811	080811620
04007		
1000064305270363028401790148		
0000050010001500200030004000		
BATCH RD FEED GANGUE 8/8/74 NO DEX TO RC	821	080811620
04007		
1000098209680953094009200908		
0000050010001500200030004000		
MINERAL BATCH RD FEED INFINITE NWTST	811	010311625
04007		
1000055005120407037403600333		
00000500100015002000300040005000		
MINERAL BATCH RD FEED INFINITE NWTST	811	080311935
04007		
1000057405120401035102870260		
00000500100015002000300040005000		
MINERAL BATCH RD FEED INFINITE NWTST	811	00140211805
04007		
1000057104940354030402360204		
0000050010001500200030004000		
MINERAL BATCH RD FEED INFINITE NWTST	811	200211810
04007		
10000484046403110254020201790104		
00000500100015002000300040005000		
MINERAL BATCH RD FEED INFINITE NWTST	811	220211920



04007  
 1000060604700310024501830156  
 00000500100015002000300040005000  
 ROUGHER FEED BATCH SULF INF I 22/2 WITH ACID 811 220211920  
 008  
 10000606047003100245018301560097  
 00000500100015002000300040004500

THESE SIZE FLOTATION CURVES ARE BUILT FROM THE COMPOSITE SAMPLES OF TESTS  
 811--010311625 811--080311935 811--140211805 811--220211920

MINERAL BATCH INFINITE +208	812	12400
04007 1000092009180873086608640863 00000500100015002000300040005000 MINERAL BATCH INFINITE +104	813	12400
04007 1000077607630648062306040601 00000500100015002000300040005000 MINERAL BATCH INFINITE +53 RF	814	12400
04007 1000072307060640059205580548 00000500100015002000300040005000 MINERAL BATCH RF INF -53	815	12400
04007 1000086707450513040403040263 00000500100015002000300040005000 COPPER BATCH RO FEED INF COMB +104	833	12400
04007 1000057805780434043404340434 0000050010001500200030004000 COPPER BATCH RO FEED INF COMB +53	834	12400
04007 1000027002570218020102010201 0000050010001500200030004000 COPPER BATCH RO FEED INF COMB -53	835	12400
04007 1000082806530497041303060256 0000050010001500200030004000 NICKEL BATCH RO FEED INF COMB +104	843	12400



04007		
1000085708570813081308130813		
0000050010001500200030004000		
NICKEL BATCH RO FEED INF COMB +53	844	12400
04007		
1000062005930526050705070507		
0000050010001500200030004000		
NICKEL BATCH RO FEED INF COMB -53	845	12400
04007		
1000089206570466040503520319		
0000050010001500200030004000		
CALC FE7S8 RO FEED +104 INF COMB	853	12400
04007		
1000081107900662061905880584		
0000050010001500200030004000		
CALC FE7S8 RO FEED +53 INF COMB	854	12400
04007		
1000082808130742068206340620		
0000050010001500200030004000		
CALC FE7S8 RO FEED -53 INF COMB	855	12400
04007		
1000086607730525040302950253		
0000050010001500200030004000		
BATCH RO CONC NO DEX 8/8/74 SULF	311	080811645
04005		
10000317019801480115		
0000050010001500200030004000		
BATCH RO CONC NO DEX 8/8/74 GANGUE	321	080811645
04005		
10000746065306060564		
0000050010001500200030004000		
BATCH RO CONC NO DEX 8/8/74 COPPER	331	080811620
04005		
10000140009200730060		
0000050010001500200030004000		
BATCH RO CONC NO DEX 8/8/74 NICKEL	341	080811620
04005		
10000247014401130093		
0000050010001500200030004000		
MINERAL BATCH RO CONC INFINITE NWTST	311	010311800

04005		
10000247016101380126		
00000500100015002000300040005000		
MINERAL BATCH RD CONC INFINITE NWTST	311	080312055
04005		
10000369016601100095		
00000500100015002000300040005000		
MINERAL BATCH RD CONC INFINITE NWTST	311	00140211930
04005		
10000220010400790069		
00000500100015002000		
MINERAL BATCH RD CONC INFINITE NWTST	311	200211945
04005		
10000251010500610046		
00000500100015002000300040005000		
MINERAL BATCH RD CONC INFINITE NWTST	311	220212045
04005		
10000242012601030091		
00000500100015002000300040005000		

THESE SIZE FLOTATION CURVES ARE BUILT UP FROM COMPOSITES OF SAMPLES FROM  
 TESTS 311--010311800 311--080312055 311--140211930 311--200211945 AND  
 311--220212045

MINERAL BATCH RC INF +208	312	12400
04005		
10000453041003970395		
00000500100015002000300040005000		
MINERAL BATCH RC INF +104	313	12400
04005		
10000439031302710269		
00000500100015002000300040005000		
MINERAL BATCH RC INF +53	314	12400
04005		
10000318017301560146		
00000500100015002000300040005000		
MINERAL BATCH RC INF -53	315	12400
04005		
10000326018500950079		
00000500100015002000300040005000		



COPPER BATCH RO CONC INF COMB +104	333	12400
04005		
10000163011901190119		
00000500100015002000		
COPPER BATCH RO CONC +53 INF COMB	334	12400
04005		
10000084005600470044		
00000500100015002000		
COPPER BATCH RO CONC INF COMB -53	335	12400
04005		
10000078004400280020		
00000500100015002000		
NICKEL BATCH RO CONC INF COMB +104	343	12400
04005		
10000369027802780278		
00000500100015002000		
NICKEL BATCH RO CONC INF COMB +53	344	12400
04005		
10000237011700990095		
00000500100015002000		
NICKEL BATCH RO CONC INF COMB -53	345	12400
04005		
10000189008600610050		
00000500100015002000		
CALC FE7S8 RO CONC +104 INF COMB	353	12400
04005		
10000621043703610356		
0000050010001500200030004000		
CALC FE7S8 RO CONC +53 INF COMB	354	12400
04005		
10000415022502060192		
00000500100015002000		
CALC FE7S8 RO CONC -53 INF COMB	355	12400
04005		
10000433025201240104		
00000500100015002000		



THESE TESTS ARE USED TO TEST THE PREDICTIVE CAPACITIES OF THE MODELS IN  
 THAT THESE BATCH TESTS WERE PERFORMED ON THE FEED TO A CONTINUOUS BANK AND  
 MAY BE USED TO PREDICT THE OPERATION OF THAT BANK \* SEE PAGES 53 AND 71

BATCH RD FEED 13/4/73 1435 INF SULF	811	130411435
04007		
1000063305650336028102440230		
0000050010001500200030004000		
GANGUE BATCH RD FEED INF 13/4/73 1435	821	130411435
04007		
1000098009670938092409110905		
0000050010001500200030004000		
ROUGHER FEED INF 13/4/73 1435 COPPER BATCH	831	130411435
04007		
1000028802450174015301340124		
0000050010001500200030004000		
ROUGHER FEED INF 13/4/73 1435 NICKEL BATCH	841	130411435
04007		
1000062604950326030502900284		
0000050010001500200030004000		
RD FEED BATCH INF 23/4/73 1200	811	230411200
04007		
1000056005180328027302480237		
0000050010001500200030004000		
GANGUE BATCH RD FEED 23/4/73 INF 1200	821	230411200
04007		
1000097109590931091609060899		
0000050010001500200030004000		
RD FEED BATCH INF 23/4/73 1200 COPPER	831	230411200
04007		
1000030102680200017401570148		
0000050010001500200030004000		
RD FEED BATCH INF 23/4/73 1200 NICKEL	841	230411200
04007		
1000048904290321029802860280		
0000050010001500200030004000		
RD FEED INF BATCH 2/5/73 1450 SULF	811	020511450
04007		
1000062805900407035503120298		

0000050010001500200030004000 RD FEED INF BATCH 2/5/73 1450 GANGUE 04007 1000097309630943093209200914 0000050010001500200030004000	821	020511450
MINERAL CL FEED BATCH 200/5 1445 11/4/73 04008 10000590040703050232014401110095 00000500100015002000300040005000	411	110411445
GANGUE BATCH CL FEED 11/4/73 1445 200/5 04008 10000894081507540711064506060571 00000500100015002000300040005000	421	110411445
CLEANER FEED BATCH 11/4/73 COPPER 1445 04008 10000270016801270106007800640053 00000500100015002000300040005000	431	110411445
CLEANER FEED BATCH 11/4/73 NICKEL 1445 04008 10000516029402050162011000880074 00000500100015002000300040005000	441	110411445
CL FEED BATCH 18/4/73 MINERAL 1150 INF 04006 100003120136009100740058 000005001000150020003000	411	180411150
GANGUE CL FEED 18/4/73 1150 INF 04006 100007290588052604920456 000005001000150020003000	421	180411150
CLEANER FEED BATCH INF 18/4/73 1150 COPPER 04006 100001950115009000780066 000005001000150020003000	431	180411150
CLEANER FEED BATCH INF 18/4/73 1150 NICKEL 04006 100002540132009600800064 000005001000150020003000	441	180411150
CLEANER FEED BATCH SULF INF 30/4/73 1430 04006 100004980309022702020182	411	300411430



000005001000150020003000

CLEANER FEED BATCH GANG INF 30/4/73 1430

421 300411430

04006

100008220711064906180583

000005001000150020003000



## \*\*\*\*\* GROUP SIX \*\*\*\*\*

THESE TESTS WERE PERFORMED IN ORDER TO TEST THE VARIOUS MODELS CAPABILITIES  
IN DESCRIBING CONTINUOUS FLOTATION \* SEE PAGES 52 AND 70

GANGUE PLANT RO FEED 47 +104	82347280921645
04011	
10000999099809970997099609950995099509940994	
00000204040906140819102312281433163818422047	
GANGUE PLANT RO FEED +53	82447280921645
04011	
10000998099609960995099509940994099409940994	
00000204040906140819102312281433163818422047	
GANGUE PLANT RO FEED 47 -53	82547280921645
04011	
10000992098209780975096909610955094009360933	
00000204040906140819102312281433163818422047	
MINERAL PLANT RO FEED 47 +208	81247280921645
04011	
10000912078007410714067806600651063306190603	
00000204040906140819102312281433163818422047	
MINERAL PLANT RO FEED 47 +104	81347280921645
04011	
10000921083908210811077907610752074307360732	
00000204040906140819102312281433163818422047	
MINERAL PLANT RO FEED 47 +53	81447280921645
04011	
10000738064806140609056505470535052305190517	
00000204040906140819102312281433163818422047	
MINERAL PLANT RO FEED 47 -53	81547280921645
04011	
10000717061805890571050204370410035903500335	
00000204040906140819102312281433163818422047	
GANGUE PLANT RO FEED 47 +208	82247280921645
04011	
10000998099609940992098909860986098309810977	
00000204040906140819102312281433163818422047	
MINERAL PLANT RUN 34 CLEANER 22/6/72	41134220621215



04021  
10000889075606860641061305980577055005390519050904960486047104590449043604260414  
0403  
0 64 129 193 258 323 387 452 517 581 646 711 775 840 905 9691034109911631228  
1293  
MINERAL PLANT CL FEED 37 41137130722115  
04021101000400700010  
10000850067105930514048204330382033003100283027302600247022902140200019001780167  
0155  
0 72 144 217 290 363 435 508 581 654 726 799 872 945101710901163123613081381  
1454  
PLANT RUN 40 9/8/72 MINERAL 41140090822000  
04021  
10000781069806240588055005180417036903470314030302860251022101930171015301400126  
0114  
0 79 159 239 319 399 479 558 638 719 799 878 9581038111811981278135714381518  
1598  
GANGUE PLANT TEST CL FEED 34 42134220621215  
04021102001103000010  
10000990097409620954094509410933092309160907090008920884087508670860084908380825  
0813  
0 64 129 193 258 323 387 452 517 581 646 711 775 840 905 9691034109911631228  
1293  
GANGUE PLANT CL FEED 37 42137130722115  
04021102001103001010  
10000978094809320916090608900874085008370822081308040793077807660754074007280714  
0699  
0 72 144 217 290 363 435 508 581 654 726 799 872 945101710901163123613081381  
1454  
PLANT RUN 40 CLE GANGUE 42140090822000  
04021  
10000962094409210906088808700818078807710742073107150683065206170591056605460522  
0499  
0 79 159 239 319 399 479 558 638 719 799 878 9581038111811981278135714381518  
1598  
PLANT RUN ND 32 MINERAL 25/5/72 8113225052  
04021  
10000878071706900660063006260610060105720566056105530541053705230517051305080502  
0499  
0 55 111 167 223 279 335 391 447 503 559 615 671 727 783 839 895 95110071063



1119

MINERAL TEST RUN 35 RD FEED PLANT

81135290621135

04021102001103000010

10000888084508260801077607560744067606500632060405830558053905270508050004940486

0479

0 45 90 136 181 227 272 318 364 409 454 500 545 591 637 682 728 773 819 864

910

MINERAL PLANT RD FEED 35

81135290621635

04021102001005011010

10000818078006770643061805990591058505640559054405360522051004900456044504360429

0425

0 46 93 140 186 233 280 326 373 420 467 513 560 607 653 700 747 794 840 887

934

MINERAL PLANT RD FEED 36

81136060721610

04021102001103001010

10000836077407610750073807280705069706040557054505330521051104950476046504630448

0443

0 46 93 141 188 235 282 330 377 424 471 521 565 612 660 707 754 801 848 896

943

MINERAL PLANT RD FEED 36

81136060722010

04021102001103001010

10000912079607890780076807560720068406280577056105330514049204660444043504310427

0415

0 45 90 135 180 226 271 316 361 407 452 498 543 588 633 678 724 769 814 859

905

MINERAL PLANT RD FEED 37

81137130721720

04021101000400700010

10000918078207440703068006580635061605730561054305310513050104780471045504480441

0428

0 62 124 187 249 311 374 436 498 561 623 685 748 810 872 935 997106011221184

1247

MINERAL PLANT RD FEED 37

81137130722050

04021101000400700010

10000903080007640700068406640646063005760551053505220477046004550435042204080394

0376

0 87 175 263 351 439 527 615 702 791 879 96610541142123013181406149415821670

1758

MINERAL PLANT RD FEED 38

81138190721615

04021102001005011010

10000830076106850663063305990587056105460493046604540446044504420441042404200415



0402

0 81 163 244 326 408 490 571 653 735 816 898 9801062114312251307138814701552

1633

MINERAL PLANT RD FEED 38

81138190722000

04021102001005011010

10000809079406760655063806010581055605350501048304730468046604620455044104300424

0416

0 87 175 263 351 438 526 614 702 789 877 96510531140122813161404149115791667

1755

MINERAL PLANT RUN 42 24/8/72

81142240821510

04021

10000839078807290687067006480635061006020601058705850581057905660557055305480543

0538

0 48 97 146 195 244 293 342 391 440 489 538 587 636 685 734 783 832 881 930

979

MINERAL PLANT RD FEED 44

81144070921800

04021101000400700010

10000914088708560827081508060796077707540735072107030690067706680631062006000591

0572

0 58 117 177 235 294 354 412 471 531 590 648 708 767 825 885 944100310621121

1180

MINERAL PLANT RD FEED 47

81147280921210

04021101000400700010

10000741072006850669065406440639062905290506049204870469046104540447044204390435

0430

0 102 204 307 409 511 614 716 819 9211023112612281330143315351638174018421945

2047

MINERAL PLANT RD FEED 47

81147280921410

04021101000400700010

10000791076807040675063606250612060405950569053205180509046704580449044204380433

0425

0 102 204 307 410 512 615 718 820 9221025112812301333143615381641174318461949

2051

MINERAL PLANT RD FEED 47

81147280921603

04021101000400700010

10000738068006550615059605880562055205400521050504910481047004640451043404290418

0409

0 112 224 336 448 561 673 785 89810101123123513471460157216841796190920212133

2246

MINERAL PLANT RD FEED 47

81147280921645



04021101000400700010

10000804077307210698068706740666065906460607057305590549053705290491048104780473  
04700 109 219 329 439 548 658 768 878 9881098120713171427153716471757186619762086  
2196

PLANT RUN 32 GANGUE 25/5/72

8213225052

04021

10000997099109890987098009790978097609740973097309720968096709650963096309610960  
09590 55 111 167 223 279 335 391 447 503 559 615 671 727 783 839 895 95110071063  
1119

GANGUE PLANT RD FEED 35

82135290621135

04021102001103001010

10000995099009880983097809740973096709650964096109570946094309400932093109300928  
09260 45 90 136 181 227 272 318 364 409 454 500 545 591 637 682 728 773 819 864  
910

GANGUE PLANT RD FEED 35

82135290621635

04021102001103001010

10000995099309870982097809750973097209700970096709650962095809540942093909380936  
09350 46 93 140 186 233 280 326 373 420 467 513 560 607 653 700 747 794 840 887  
934

GANGUE PLANT RD FEED 36

82136060721610

04021102001103001010

10000993098909870986098209810978097609460940093809370935093309310927092509250922  
09210 46 93 141 188 235 282 330 377 424 471 521 565 612 660 707 754 801 848 896  
943

GANGUE PLANT RD FEED 36

82136060722010

04021102001103001010

10000997099309920991098809870981096609530946094509420937092909230919091709160915  
09140 45 90 135 180 226 271 316 361 407 452 498 543 588 633 678 724 769 814 859  
905

GANGUE PLANT RD FEED 37

82137130721720

04021102001103001010

10000997099309900986098109780976097009660964096309610959095609510950094809450944  
0942

0 62 124 187 249 311 374 436 498 561 623 685 748 810 872 935 997106011221184



1247

GANGUE PLANT RD FEED 37

82137130722050

04021102001103001010

100009960991098709790976097409720969096209590957095509450941094009340932092709240920

0 87 175 263 351 439 527 615 702 791 879 966105411421230131814061494158216701758

GANGUE PLANT RD FEED 38

82138190721615

04021102001103001010

100009910989098209790968095009480944094409410939093809370937093609360933093309310929

0 81 163 244 326 408 490 571 653 735 816 898 98010621143122513071388147015521633

GANGUE PLANT RD FEED 38

82138190722000

04021102001103001010

100009960995098909870980096809630959095709560953095209510951094909490946094409420940

0 87 175 263 351 438 526 614 702 789 877 965105311401228131614041491157916671755

GANGUE PLANT RUN 42 24/8/72

82142240821510

04021

100009940991098709810979097509710965096509650964096409630963096209610961096009590959

0 48 97 146 195 244 293 342 391 440 489 538 587 636 685 734 783 832 881 930979

GANGUE PLANT RD FEED 44

82144070921800

04021102001103001010

100009980997099509910990099009880986098509830982098009790978097709700969096609650962

0 58 117 177 235 294 354 412 471 531 590 648 708 767 825 885 9441003106211211180

GANGUE PLANT RD FEED 47

82147280921210

04021101000501401010

100009930992099009870986098409830981097809760974097409710970096809670965096409630962

0 102 204 307 409 511 614 716 819 92110231126122813301433153516381740184219452047

GANGUE PLANT RD FEED 47

82147280921410

04021101000501401010

1000099609950992098709840983098009790977097409720971095909580956095409540953



0951  
0 102 204 307 410 512 615 718 820 9221025112812301333143615381641174318461949  
2051  
GANGUE PLANT RO FEED 47 82147280921603  
04021101000501401010  
10000994099109890985098309820974097309720971096909680967096509640962095909590957  
0955  
0 112 224 336 448 561 673 785 89810101123123513471460157216841796190920212133  
2246  
GANGUE PLANT RO FEED 47 82147280921645  
04021101000501401010  
1000099609950993099098909870986098509840982097909780976097509730967096509650964  
0964  
0 109 219 329 439 548 658 768 878 9881098120713171427153716471757186619762086  
2196  
COPPER PLANT RO FEED 47 1410 83147280921410  
04021  
10000531050204330408039103810378036203590352034303390329030403000296029402920289  
0287  
0 102 204 307 410 512 615 718 820 9221025112812301333143615381641174318461949  
2051  
NICKEL PLANT RO FEED 47 1410 84147280921410  
04021  
10000795076907110674064606330630061406100600058705800572054105350530052605240521  
0519  
0 102 204 307 410 512 615 718 820 9221025112812301333143615381641174318461949  
2051

THE FOLLOWING TESTS ON A CONTINUOUS BANK OF CELLS HAVE COMPLEMENTARY BATCH  
TEST DATA ON THEIR FEED STREAMS \* SEE PAGES 53 AND 71

ROUGHER A3 13/4/73 1435 MINERAL

811 130421435

04021

10000912076506160578056705580538051304430417040904010388038203760360035103400331

0321

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

ROUGHER A3 13/4/73 1435 GANGUE

821 130421435

04021

1000099809930987098409830982097909750971096909680967096609640962096109580957

0956

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

ROUGHER BANK COPPER 13/4/72 1430

831 130421430

04021

10000695047903730354034603420331031702940282027702740268026202590250024502360232

0226

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

ROUGHER BANK NICKEL 13/4/72 1430

841 130421430

04021

10000934076306280597058205730557053705160506050305000495049204910486048204780476

0472

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

ROUGHER A3 23/4/73 1200 MINERAL

811 230421200

04021

10000767063305210510049804940491048004010372035803510317030702950274026902590250

0237

0 68 137 205 274 343 411 480 549 618 686 755 824 892 96110291098116712361304

1373

ROUGHER A3 23/4/73 1200 GANGUE

821 230421200

04021

10000994098709780977097509740973097009640962096009590952095009480941094009370935

0931

0 68 137 205 274 343 411 480 549 618 686 755 824 892 96110291098116712361304

1373



ROUGHER BANK 2304/73 1200 COPPER 831 230421200  
04021  
10000600044602490241023202280226022002040198019201900177017201680156015201480144  
0139  
0 68 137 205 274 343 411 480 549 618 686 755 824 892 96110291098116712361304  
1373  
ROUGHER BANK 2304/73 1200 NICKEL 841 230421200  
04021  
10000750056504600449043604310428041804010395039003880379037503720363036103580355  
0351  
0 68 137 205 274 343 411 480 549 618 686 755 824 892 96110291098116712361304  
1373  
ROUGHER TEST 2/5/73 1450 SULFUR 811 020521450  
04021  
10000777072405850498045304280424040403980384038103690327032002980277027202660261  
0256  
0 68 137 205 274 342 411 479 548 616 685 753 822 890 95910271096116412331301  
1370  
ROUGHER TEST 2/5/73 1450 GANGUE 821 020521450  
04021  
10000995099309870976097109700969096509640963096309620957095609530950094809470946  
0945  
0 68 137 205 274 342 411 479 548 616 685 753 822 890 95910271096116412331301  
1370  
CLEANER 11/4/73 A SECT 1445 SULF 411 110421445  
04021  
10000862080807520642052404740438039203570329030702710239021701990173016901680167  
0162  
0 97 195 293 390 488 587 684 782 880 978107611741272136914671566166317611859  
1957  
CLEANER A SECTION 11/4/73 1445 GANGUE 421 110421445  
04021  
10000978097009590937090408840868084608220801078307560733071206930668066306610659  
0650  
0 97 195 293 390 488 587 684 782 880 978107611741272136914671566166317611859  
1957  
CLEANER BANK 11/4/73 1445 COPPER 431 110421445  
04021  
10000681057804790349025202130189016801500138012801160104009500870078007600750074  
0070



0 97 195 293 390 488 587 684 782 880 978107611741272136914671566166317611859  
1957

CLEANER BANK 11/4/73 1445 NICKEL 441 110421445  
04021

10000869081407420590043403710326028102460222020401810163014901370123012001180117  
0112

0 97 195 293 390 488 587 684 782 880 978107611741272136914671566166317611859  
1957

CLEANER A SECTION 18/4/73 MINERAL 411 180421145  
04021

10000902081207660720066906370602055804990435039503380306028102710259024602320225  
0214

0 87 175 263 350 438 526 614 702 790 877 96510531141122813161405149215801668  
1755

CLEANER A SECT 18/4/73 GANGUE 1145 421 180421145  
04021

10000984096309490935091809060894087808550830081307870768075307450737072607140708  
0696

0 87 175 263 350 438 526 614 702 790 877 96510531141122813161405149215801668  
1755

CLEANER BANK 18/4/73 1145 COPPER 431 180421145  
04021

10000704054804860444041603980384036703450320030502840266025302470241023402260222  
0217

0 87 175 263 350 438 526 614 702 790 877 96510531141122813161405149215801668  
1755

CLEANER BANK 18/4/73 1145 NICKEL 441 180421145  
04021

10000894073706600584051704760436039903590320029602670247023302260217020901990194  
0186

0 87 175 263 350 438 526 614 702 790 877 96510531141122813161405149215801668  
1755

CLEANER BANK 30/4/73 1430 SULFS 411 300421430  
04021

10000829071906370582050704200366031602690242022502040203020102000195018901840176  
0164

0 125 251 376 502 629 754 880100611311258138315091635176018872013213822642390  
2516

CLEANER BANK 30/4/73 1430 GANGUE 421 300421430  
04021

10000957092308890862082307800747071306950671065306260623062106190610060005930583  
0567

0 125 251 376 502 629 754 880100611311258138315091635176018872013213822642390  
2516



## \*\*\*\*\* GROUP SEVEN \*\*\*\*\*

THESE SPECIAL TESTS ARE USED IN ANALYZING THE FROTH PHASE OF FLOTATION \*  
SEE PAGE 105

MINERAL BATCH RD CONC 17-3-74 FROTH TEST 15 G/MIN	311	170311320
007		
1000081607260636055104180311		
0000050010001500200030004000		
GANGUE BATCH RD CONC 17-3-74 FROTH TEST 15 G/MIN	321	170311320
007		
1000097909660952093909160892		
0000050010001500200030004000		
MINERAL BATCH RD CONC 17-3-74 FROTH TEST 60G/MIN	311	170311500
007		
1000063004900386030702200159		
0000050010001500200030004000		
GANGUE BATCH RD CONC 17-3-74 FROTH TEST 60G/MIN	321	170311500
007		
1000094409080877085008040765		
0000050010001500200030004000		
MINERAL BATCH RD CONC 17-3-74 FROTH TEST 30G/MIN	311	170311610
007		
1000074706000496040502920232		
0000050010001500200030004000		
GANGUE BATCH RD CONC 17-3-74 FROTH TEST 30G/MIN	321	170311610
007		
1000096809440925090708770854		
0000050010001500200030004000		
MINERAL BATCH RD CONC 17-3-74 FROTH TEST 120G/MIN	311	170311715
006		
100005640410030802480171		
0000050010001500200030004000		
GANGUE BATCH RD CONC 17-3-74 FROTH TEST 120G/MIN	321	170311715
006		
100009070849080307660701		
0000050010001500200030004000		
RD CONC 15G/MIN BATCH 1320	331	170311320
COPPER		



007	
1000048103850270020401310109	
0000050010001500200030004000	
BATCH RD CONC NICKEL 15 G/MIN	341 170311320
007	
1000089108190742063403360150	
0000050010001500200030004000	
BATCH RD CONC COPPER 60 G/MIN	331 170311500
007	
1000033601730114008300530043	
0000050010001500200030004000	
BATCH RD CONC NICKEL 60 G/MIN	341 170311500
007	
1000068705070317018500960059	
0000050010001500200030004000	
BATCH RD CONC COPPER 30 G/MIN	331 170311610
007	
1000049102670187014100900067	
0000050010001500200030004000	
BATCH RD CONC NICKEL 30 G/MIN	341 170311610
007	
1000082806650485032401430084	
0000050010001500200030004000	
BATCH RD CONC COPPER 120 G/MIN	331 170311715
006	
100002610177011300960059	
0000050010001500200030004000	
BATCH RD CONC NICKEL 120 G/MIN	341 170311715
006	
100005190326018801280063	
0000050010001500200030004000	

```

// FOR FTSRT           STARTS ANYFT
*IOCS(CARD)
C
C** PROGRAM FOR INITIALIZATION OF REGRESSION OF BATCH OR CONTINUOUS TEST DATA
C BY THE DERIVATIVE METHOD DESCRIBED ON PAGES 142-144
C** ICL - AN ARRAY OF SEVEN WHICH SPECIFIES WHICH TEST CODES ARE TO BE REGRESSED
C IF ICL IS ZERO ALL TESTS INPUT WILL BE REGRESSED.
C** IWT - AN ARRAY OF FIVE USED TO SPECIFY WHICH MODEL FORM IS TO BE REGRESSED.
C HOLDS A VALUE OF EITHER ZERO OR UNITY. THE GENERAL FORM OF THE BATCH TEST
C EQUATION IS  $C = (IWT1 * A * \exp(-100 * T) + (1 - IWT1 * A - IWT2 * B) * \exp(-IWT3 * C * T ** D - IWT4 * C * T - IWT5 * T ** D) + IWT2 * B)$  WHERE A,B,C,D ARE POSSIBLE PARAMETERS. THIS MAY BE
C SIMPLIFIED IF ONLY THE SINGLE RATE AND RATE PLUS NONFLOATING MODELS ARE OF
C INTEREST. SEE ALSO SUBROUTINES THEO AND DERIV.
C** A SIMILAR CONTINUOUS EQUATION IS USED.  $C = (1 - IWT1 * A - IWT2 * B) / (1 + C * IWT3 * T) ** N + IWT2 * B$  WHERE A,B,C ARE POSSIBLE PARAMETERS.
C
      EXTERNAL SETUP
      COMMON Y(21),THEOR(21),SIQSG(21),T(21),RES(21),PF(4),PPF(4,4),P(4)
      1,X(4),C(4),A(4,4),ITIT(30),NPAR,NDATA,ITMAX,NCSTP,NCTGT,CHMIN,IWT(
      25),CHISQ,K,ID(7),ILC(7),LC
      K=0
C
C** FREE FORMAT CARD INPUT TO ARRAY Y
C
      CALL RDCRD (13,MN,Y)
      DO 1 L=1,7
1       ILC(L)=Y(L)
      DO 2 L=1,5
2       IWT(L)=Y(L+7)
      LC=Y(13)
      LC=1
      CALL LINK (SETUP)
      END

```



```

// FOR SETUP
*IOCS(CARD)
C
C** READS IN BATCH/CONTINUOUS TEST DATA AND DESCRIPTION FOR ANYFT DERIVATIVE
C REGRESSION PROGRAM.
C** CALCULATES INITIAL PARAMETER ESTIMATES AND CHECKS TEST CODE TO SEE IF TEST
C IS TO BE REGRESSED.
C** LINKED TO BY FTSRT OR ANYFT. LINKS TO BLOK1.
C
  EXTERNAL BLOK1
  COMMON Y(21),THEOR(21),SIQSG(21),T(21),RES(21),PF(4),PPF(4,4),P(4)
  1,X(4),C(4),A(4,4),ITIT(30),NPAR,NDATA,ITMAX,NCSTP,NCTCT,CHMIN,IWT(
  25),CHISQ,JL,ID(7),ILC(7),LC
  DO 7 M=1,400
    READ (2,500) ITIT,ID,NDATA
500  FORMAT (30A2,6X,3I1,I2,I4,I1,I4/2X,I3)
    READ (2,501) (Y(I),I=1,NDATA)
501  FORMAT (20F4.3)
    READ (2,502) (T(I),I=1,NDATA)
502  FORMAT (20F4.2)
    NIT=0
    NITC=0
    DO 8 K=1,7
      IF (ILC(K)) 8,8,9
9    NIT=NIT+1
      IF (ILC(K)-ID(K)) 8,10,8
10   NITC=NITC+1
8    CONTINUE
      IF (NIT-NITC) 7,14,7
7    CONTINUE
      CALL EXIT
14   GO TO (40,41),LC
41   DO 42 I=1,NDATA
42   T(I)=SQRT(T(I))
40   ITMAX=9
      NCSTP=3
      NCTCT=8
      NPAR=4
      P(1)=(1.-Y(2)+.05)*IWT(1)
      P(2)=(Y(NDATA)-.05)*IWT(2)

```



```
TTT=0.
SSS=0.
XX1=Y(1)-IWT(1)*P(1)-IWT(2)*P(2)
DO 73 I=1,NDATA
XX0=Y(I)          -IWT(2)*P(2)
SSS=ALOG(XX0/XX1)+SSS
73  TTT=TTT-T(I)
    P(3)=SSS/TTT
    CALL LINK (BLOK1)
END
```

```

// FOR BLOK1
*IOCS(1443 PRINTER)
C
C** DATA OUTPUT FOR BATCH/CONTINUOUS TEST REGRESSIONS FOR SINGLE RATE OR RATE
C PLUS NONFLOATING FIRST ORDER MODELS. LINKED TO BY SETUP. LINKS TO ANYFT.
C
COMMON DATA(21),THEOR(21),SIGSQ(21),ABCIS(21),RES(21),PF(4),PPF(4,
14),PAR(4),OLD(4),C(4),A(4,4),ITIT(30),NPAR,NDATA,ITMAX,NCSTP,NCTCT
2,CH4IN,IWT(5),CHISQ,KH,ID(7)
WRITE(3,3) ITIT,ID
WRITE(3,4) NPAR
DO 5 I=1,NPAR
WRITE (3,6) I,PAR(I)
5 CONTINUE
WRITE (3,7) ITMAX,NCSTP,NCTCT
WRITE (3,8) NDATA,(DATA(I),I=1,NDATA)
WRITE(3,9) (ABCIS(I),I=1,NDATA)
C
C** MAY BE ALTERED TO WEIGHT INPUT DATA FOR REGRESSION
10 DO 11 I=1,NDATA
11 SIGSQ(I)=1.
C
WRITE(3,12)
45 CHISQ=10.**10
WRITE(3,25) ITIT
NIT=0
NCUT=0
NCUTT=0
CALL LINK (ANYFT)
3 FORMAT (1H1,30A2,7I5)
4 FORMAT (' THE FOLLOWING SUPPLIED BY USER   '/ 20X,' NUMBER OF PAR
1AMETERS = ',I3)
6 FORMAT (20X,'INITIAL VALUES FOR PARAMETER NO. ',I4,' IS ',E16.8)
7 FORMAT (20X,'MAX NO. OF ITERATIONS ALLOWED',I5/20X,'MAX NO. OF CON
1SECUTIVE CUT STEPS ',I5/20X,'MAX TOTAL NO. OF CUT STEPS ',I5)
8 FORMAT ( /20X,'THE',I4,' DATA POINTS ARE'/20(5E20.8/))
9 FORMAT ( /20X,'THE CORRESPONDING ABCISSA ARE'/20(5E20.8/))
12 FORMAT (20X,' STANDARD DEVIATION = 1. FOR EACH')
25 FORMAT (/20X,' ITERATION SEQUENCE FOR ',30A2/, ' ITERATION
1CUT CHI SQUARED P A R A M E T E R S ',/)
END

```



```

// FOR ANYFT
*IOCS(1443 PRINTER)
C
C** MAIN REGRESSION PROGRAM FOR BATCH/CONTINUOUS TEST DATA FOR USE WITH SINGLE
C RATE AND RATE PLUS NONFLOATING MODELS. LINKS TO SETUP.
C** DATA - CONTAINS OBSERVED SPECIES CONCENTRATIONS
C** THOER - CONTAINS CALCULATED SPECIES CONCENTRATIONS
C** SIGSQ - NOT USED
C** ABCIS - CONTAINS FLOTATION TIMES CORRESPONDING TO SPECIES CONCENTRATIONS
C** RES - NOT USED
C** PF - CONTAINS FIRST DERIVATIVE OF THOER WITH RESPECT TO EACH PARAMETER.
C** PPF - CONTAINS SECOND ORDER DERIVATIVES
C** PAR - CONTAINS PRESENT PARAMETER VALUES OLD - CONTAINS PREVIOUS PARAMETER
C VALUES.
C** C,A - MATRICES USED IN THE REGRESSION
C** ITIT - A 60 CHARACTER TEST DESCRIPTION
C** NPAR - NUMBER OF PARAMETERS TO BE REGRESSED
C** NDATA - NUMBER OF OBSERVED CONCENTRATIONS
C** ITMAX - MAXIMUM NUMBER OF ITERATIONS TO EXIT
C** NCSTP - MAXIMUM NUMBER OF CUT STEPS PER ITERATION
C** NCTCT - MAXIMUM NUMBER OF CUT STEPS TO EXIT
C** CHMIN - PROGRAM EXITS ON A SUM OF SQUARES LESS THAN CHMIN
C** IWT - MODEL CODE
C** CHISQ - PRESENT SUM OF SQUARES
C** ID - TEST IDENTITY CODE
C
      EXTERNAL SETUP
      COMMON DATA(21),THOER(21),SIGSQ(21),ABCIS(21),RES(21),PF(4),PPF(4,
14),PAR(4),OLD(4),C(4),A(4,4),ITIT(30),NPAR,NDATA,ITMAX,NCSTP,NCTCT
2,CHMIN,IWT(5),CHISQ,KH,ID(7)
      CHMIN=0.
54   PREV=CHISQ
C
C** CALL TO SUBROUTINE TO GENERATE THEORETICAL MODEL VALUES
C
55   CALL THEO
      CHISQ=0
      DO 60 I=1,NDATA
60   CHISQ=CHISQ+(THOER(I)-DATA(I))**2
      WRITE (3,70) NIT,NCUT,CHISQ,(PAR(I),I=1,NPAR)

```



```

70  FORMAT (6X,I5,I8,6E16.7/35X,5E16.7/)
    IF(CHISQ-PREV) 100,1003,1003
1003 IF(NIT) 57,100,57
57  NCUT=NCUT+1
    IF(NCUT-NCSTP) 1004,1004,100
1004 NCUTT=NCUTT+1
    IF(NCUTT-NCTCT) 1005,1005,900
1005 DO 81 I=1,NPAR
81  PAR(I)=OLD(I)+(PAR(I)-OLD(I))*0.5
    GO TO 55
100  IF(CHISQ-CHMIN) 900,1006,1006
1006 NIT=NIT+1
    NCUT=0
    DO 101 I=1,NPAR
    C(I)=0.
    DO 101 J=1,NPAR
    A(I,J)=0.
101  CONTINUE
    DO 110 I=1,NDATA
C
C** CALL TO SUBROUTINE TO GENERATE FIRST AND SECOND PARTIAL DERIVATIVES
C
    CALL DERIV (I)
    DO 110 M=1,NPAR
    C(M)=C(M)+2.*((THDER(I)-DATA(I))*PF(M)
    DO 110 J=1,NPAR
    A(J,M)=A(J,M)+(2.*((THDER(I)-DATA(I))*PPF(J,M)+PF(J)*PF(M
1)))
110  CONTINUE
C
C** CALL TO SUBROUTINE FOR DIRECT MATRIX INVERSION
C
    CALL MINV5 (A,NPAR)
1007 IF(NIT-ITMAX) 1008,1008,900
1008 DO 120 K=1,NPAR
    X=0.
    DO 121 KK=1,NPAR
121  X=X+A(K,KK)*C(KK)
    OLD(K)=PAR(K)
120  PAR(K)=PAR(K)-X

```

900 GO TO 54  
CALL LINK (SETUP)  
END

```
// FOR THEO          VAR FOR PLANT
SUBROUTINE THEO
```

```
C
C** CALCULATES THE THEORETICAL CONTINUOUS TEST CONCENTRATIONS FROM REGRESSED
C PARAMETERS. USED BY ANYFT.
C
```

```
COMMON Y(21),THEOR(21),SIQSG(21),T(21),RES(21),PF(4),PPF(4,4),P(4)
1,X(4),C(4),A(4,4),ITIT(30),NPAR,NDATA,ITMAX,NCSTP,NCTCT,CHMIN,IWT(
25),CHISQ,JL,ID(7),ILC(7),LC
```

```
P(1)=P(1)*IWT(1)
```

```
P(2)=P(2)*IWT(2)
```

```
P(3)=P(3)*IWT(4)
```

```
THEOR(1)=1.
```

```
DO 1 I=2,NDATA
```

```
1 THEOR(I)=(1.-P(1)*IWT(1)-P(2)*IWT(2))/(1.+P(3)*T(2))**(I-1)+P(2)*I
WT(2)
```

```
RETURN
```

```
END
```



```

// FOR DERIV
  SUBROUTINE DERIV(K)
C
C** CALCULATES THE FIRST AND SECOND ORDER ANALYTICAL PARTIAL DERIVATIVES OF THE
C THEORETICAL CONTINUOUS TEST CONCENTRATIONS WITH RESPECT TO DESIGNATED
C PARAMETERS. USED BY ANYFT.
C
  COMMON Y(21),THEOR(21),SIQSG(21),T(21),RES(21),PF(4),PPF(4,4),P(4)
  1,X(4),C(4),A(4,4),ITIT(30),NPAR,NDATA,ITMAX,NCSTP,NCTCT,CHMIN,IWT(
  25),CHISQ,JL,ID(7),ILC(7),LC
  P(1)=P(1)*IWT(1)
  P(2)=P(2)*IWT(2)
  P(3)=P(3)*IWT(4)
  DO 38 J=1,4
  PF(J)=0.
  DO 38 L=1,4
38 PPF(J,L)=0.
  I=K-1
  PF(1)=-1./(1.+P(3)*T(2))**I*IWT(1)
  PF(2)=(PF(1)+1.)*IWT(2)
  PF(3)=-I*T(2)*(1.-P(1)-P(2))/(1.+P(3)*T(2))**(I+1)
  PPF(1,3)=I*T(2)/(1.+P(3)*T(2))**K
  PPF(2,3)=PPF(1,3)
  PPF(3,1)=PPF(1,3)
  PPF(3,2)=PPF(1,3)
  PPF(3,3)=I*T(2)**2.*K*(1.-P(1)-P(2))/(1.+P(3)*T(2))**(I+2)
  PPF(1,3)=PPF(1,3)*IWT(1)
  PPF(2,3)=PPF(2,3)*IWT(2)
  PPF(3,1)=PPF(3,1)*IWT(1)
  PPF(3,2)=PPF(3,2)*IWT(2)
  RETURN
  END

```

```
// FOR DERIV
```

```
C
```

```
C** CALCULATES THE FIRST AND SECOND ORDER ANALYTICAL PARTIAL DERIVATIVES OF THE  
C THEORETICAL BATCH TEST CONCENTRATIONS WITH RESPECT TO THE DESIGNATED MODEL  
C PARAMETERS. USED BY ANYFT.
```

```
C
```

```
      SUBROUTINE DERIV(I)
      COMMON Y(21),THEOR(21),SIQSG(21),T(21),RES(21),PF(4),PPF(4,4),P(4)
      1,X(4),C(4),A(4,4),ITIT(30),NPAR,NDATA,ITMAX,NCSTP,NCTCT,CHMIN,IWT(
      25),CHISQ
      Z=-IWT(3)*P(3)*T(I)**P(4)-IWT(4)*P(3)*T(I)-IWT(5)*T(I)**P(4)
      R=(1.-IWT(1)*P(1)-IWT(2)*P(2))
      W=(-IWT(3)*T(I)**P(4)-IWT(4)*T(I))
      V=(-IWT(3)*P(3)*T(I)**P(4)*ALOG(T(I))-IWT(5)*T(I)**P(4)*ALOG(T(I))
      1)
      PF(1)=IWT(1)*EXP(-100.*T(I))-IWT(1)*EXP(Z)
      PF(2)=-IWT(2)*EXP(Z)+IWT(2)
      PF(3)=R*W*EXP(Z)
      PF(4)=R*V*EXP(Z)
      PPF(1,1)=0.
      PPF(2,2)=0.
      PPF(3,3)=R*W**2*EXP(Z)
      PPF(4,4)=R*V**2*EXP(Z)+R*(-IWT(3)*P(3)*T(I)**P(4)*(ALOG(T(I)))**2-
      1 IWT(5)*T(I)**P(4)*(ALOG(T(I)))**2)*EXP(Z)
      PPF(1,2)=0.
      PPF(1,3)=-IWT(1)*W*EXP(Z)
      PPF(1,4)=-IWT(1)*V*EXP(Z)
      PPF(2,1)=0.
      PPF(2,3)=-IWT(2)*W*EXP(Z)
      PPF(2,4)=-IWT(2)*V*EXP(Z)
      PPF(3,1)=PPF(1,3)
      PPF(3,2)=PPF(2,3)
      PPF(3,4)=R*(-IWT(3)*T(I)**P(4)*ALOG(T(I)))*EXP(Z)+R*W*V*EXP(Z)
      PPF(4,1)=PPF(1,4)
      PPF(4,2)=PPF(2,4)
      PPF(4,3)=PPF(3,4)
      RETURN
      END
```



```

// FOR THEO
C
C** CALCULATES THEORETICAL MODEL VALUES FOR THE BATCH CASE OF THE SINGLE RATE
C   AND RATE PLUS NONFLOATING MODEL. USED BY PROGRAM ANYFT.
C** ARRAY THEOR CONTAINS THE THEORETICAL VALUES AS NORMALIZED CONCENTRATIONS.
C
      SUBROUTINE THEO
      COMMON Y(21),THEOR(21),SIQSG(21),T(21),RES(21),PF(4),PPF(4,4),P(4)
      1,X(4),C(4),A(4,4),ITIT(30),NPAR,NDATA,ITMAX,NCSTP,NCTCT,CHMIN,IWT(
      25),CHISQ
      DO 1 I=1,NDATA
1      THEOR(I)=IWT(1)*P(1)*EXP(-100.*T(I))+(1.-IWT(1)*P(1)-IWT(2)*P(2))*
      1EXP(-IWT(3)*P(3)*T(I)**P(4)-IWT(4)*P(3)*T(I)-IWT(5)*T(I)**P(4))+IW
      2T(2)*P(2)
      RETURN
      END

```



// XEQ FSTRT FX1

0 0 0 0 0 0 0 1 0 1 0

GANGUE PLANT RO FEED 47 +104

82347280921645

04011

1000099909980997099709960995099509940994

00000204040906140819102312281433163818422047

GANGUE PLANT RO FEED +53

82447280921645

04011

1000099809960996099509950994099409940994

00000204040906140819102312281433163818422047

GANGUE PLANT RO FEED 47 -53

82547280921645

04011

10000992098209780975096909610955094009360933

00000204040906140819102312281433163818422047

MINERAL PLANT RO FEED 47 +208

81247280921645

04011

10000912078007410714067806600651063306190603

00000204040906140819102312281433163818422047

MINERAL PLANT RO FEED 47 +104

81347280921645

04011

10000921083908210811077907610752074307360732

00000204040906140819102312281433163818422047

MINERAL PLANT RO FEED 47 +53

81447280921645

04011

10000738064806140609056505470535052305190517

00000204040906140819102312281433163818422047

MINERAL PLANT RO FEED 47 -53

81547280921645

04011

10000717061805890571050204370410035903500335

00000204040906140819102312281433163818422047

GANGUE PLANT RO FEED 47 +208

82247280921645

04011

10000998099609940992098909860986098309810977

00000204040906140819102312281433163818422047

```

// FOR REGAL                ALF,CN BATCH REG ** VA02A
*IOCS(CARD)
*IOCS(TYPEWRITER)
C
C** ALTERNATE PROGRAM FOR THE REGRESSION OF BATCH DATA FOR THE RATE PLUS
C  NONFLOATING MODEL USING THE POWELL SEARCH METHOD.
C** IS,ID - TEST IDENTIFICATION
C** E - PARAMETER TOLERANCES
C** ZZZ - WORKING ARRAY FOR VA02A
C** Y - OBSERVED SPECIES CONCENTRATIONS
C** T - FLOTATION RETENTION TIMES
C** P - MODEL PARAMETERS
C** PRED - BEST FIT SPECIES CONCENTRATIONS
C
      DIMENSION IS(30),ID(7)
      DIMENSION E(3),ZZZ(100),Y(13),T(13),P(3),PRED(13)
      DATA E / .001, .005, .1 /
2     CALL BREAD (T,Y,IS,ID,NDATA)
      P(2)=Y(NDATA)-.02
      XX=0.
      X=0.
      DO 1 I=1,NDATA
      X=X+ALOG(Y(I)-P(2))
1     XX=XX+T(I)
      P(1)=-X/XX
      NNM=2+3*(NDATA+2*3/2)
      ES=5.
      CALL VA02A (NDATA,2,PRED,P,E,ES,-20,200,ZZZ,NNM,Y,T)
      W=0.
      DO 302 I=1,NDATA
      PRED(I)=(1.-P(2))*EXP(-P(1)*T(I))+P(2)
302    W=W+(Y(I)-PRED(I))**2.0
      WRITE (3,100) IS,ID,NDATA
100    FORMAT (30A2,8I5)
      WRITE (3,103) P,W,(Y(I),T(I),PRED(I),I=1,NDATA)
103    FORMAT (4E12.4/13(3F10.4//))
      GO TO 2
99    CALL EXIT
      END

```



```
// FOR MODSQ          ALF,CN BATCH REG
      SUBROUTINE MODSQ (FF,M,N,P,F,TT)
C
C** SUBROUTINE FOR USE WITH POWELL SEARCH (VA02A) REGRESSION OF BATCH TEST
C   PARAMETERS OF THE RATE PLUS NONFLOATING MODEL
C
      DIMENSION FF(1),P(3),F(1),TT(1)
      DO 1 I=1,M
      XXX=(1.-P(2))*EXP(-P(1)*TT(I))+P(2)
1     F(I)=FF(I)-XXX
      RETURN
      END
```



// XEQ REGAL FX

MINERAL VOLLY FLOT FEED 23/3/73 4/3

111 23031

04011

10000706056004580380033402980266024202220207

00000500100015002000250030003500400050006000

GANGUE VOLLY FLOT FEED 23/3/73 4/3

121 23031

04011

10000990098309780972096809660960095609520949

00000500100015002000250030003500400050006000

COPPER BATCH VOLLY FLOT FEED 4/3 23-3-73

131 23031

011

10000432030702520220020201880176016601560149

00000500100015002000250030003500400050006000

NICKELR BATCH VOLLY FLOT FEED 4/3 23-3-73

141 23031

011

10000792059604840417038403620343033103220314

00000500100015002000250030003500400050006000

FLOT FEED BATCH NON CU MAT 23/3/73

161 23071

011

10000989098209750969096509610956095209490946

00000500100015002000250030003500400050006000

FLOT FEED BATCH NON NI MAT 23/3/73

171 23071

011

10000989098209740970096609620957095409500946

00000500100015002000250030003500400050006000

```

// FOR REGDL
*IOCS(CARD)
*IOCS(TYPEWRITER)
    EXTERNAL DIST
    DIMENSION E(3)
    DIMENSION ZZZ(100)
    DIMENSION Y(13),T(13),P(3),FLIM(4),TT(20),PRED(13)
    DATA FLIM /-.9,20.,0.,300./
    DATA E/.01,.05,.001/
C** MAINLINE PROGRAM FOR THE REGRESSION OF BATCH FLOTATION DATA TO FIT THE GAMMA
C  FUNCTION MODEL.
C** THE POWELL SEARCH METHOD IS USED TO GENERATE MEAN AND VARIANCE OF THE BEST
C  FITTING RATE CONSTANT DISTRIBUTION FOR BOTH CASES 1) ZERO MATERIAL IN THE
C  SPECIES IS NONFLOATABLE 2) Y(NDATA) OR LAST OBSERVED FRACTION OF THE SPECIES
C  IS CONSIDERED TO BE NONFLOATABLE.
C** E - PARAMETER TOLERANCES FOR EXIT FROM POWELL SEARCH
C** ZZZ - WORKING ARRAY FOR VAO2A
C** Y - OBSERVED NORMALIZED SPECIES CONCENTRATION FROM BATCH TEST
C** T - BATCH TEST RETENTION TIMES
C** P - PARAMETER DISTRIBUTION VALUES
C** FLIM - PARAMETER SEARCH LIMITS FOR CASE TWO GRID SEARCH
C** TT - TEST IDENTIFICATION AND CODE
C** PRED - CALCULATED SPECIES CONCENTRATIONS FROM BEST FIT PARAMETERS
C** A - ON OUTPUT IS MEAN OF RATE DISTRIBUTION
C** B - ON OUTPUT IS VARIANCE OF RATE DISTRIBUTION
C** W - SUM OF SQUARES AT BEST FIT
C** NDATA - NUMBER OF OBSERVED CONCENTRATIONS
107  FORMAT (' STAGE ONE ',3E12.4)
9    READ (2,100) TT,NDATA
100  FORMAT (20A4/2X,I3)
    READ (2,101) (Y(I),I=1,NDATA)
101  FORMAT (20F4.3)
    READ (2,102) (T(I),I=1,NDATA)
102  FORMAT (20F4.2)
C
C** GET FIRST GUESS PARAMETER VALUES AS DESCRIBED ON PAGES 144-145
C
    W=10.**10.
    P(3)=0.
    DEL=10.

```



```

      P(2)=-9.99
      DO 300 L=1,4
      DO 400 K=1,10
      P(2)=P(2)+DEL
      X=0.
      XX=0.
      DO 2 J=2,NDATA
      X=X+ALOG(Y(J))*ALOG(P(2)/(P(2)+T(J)))
      XX=XX+(ALOG(P(2)/(P(2)+T(J))))**2.
      P(1)=X/XX-1.0
      CALL DIST1(PRED,NDATA,T,P)
      S=0.
      DO 3 J=1,NDATA
      S=S+(Y(J)-PRED(J))**2.
      IF(S-W) 401,402,402
401  X1=W
      W=S
      A=P(1)
      B=P(2)
      GO TO 400
402  P(2)=P(2)-2.*DEL
      IF(P(2)) 404,405,405
404  P(2)=.01
405  DEL=DEL/5.
      W=X1
      GO TO 300
400  CONTINUE
300  CONTINUE
      P(1)=A
      P(2)=B
      ES=10.
      NNM=2+3*(NDATA+2*3/2)
C
C** INITIAL GUESSES AVAILABLE, BEGIN POWELL SEARCH FOR BEST FIT
C
      CALL VAO2A (NDATA,2,PRED,P,E,ES,-5,100,ZZZ,NNM,Y,T)
      CALL DIST1 (PRED,NDATA,T,P)
      W=0.
      DO 301 I=1,NDATA
301  W=W+(Y(I)-PRED(I))**2.0

```



```

      A=(P(1)+1.)/P(2)
      B=(P(1)+1.)/P(2)**2.
C
C** OUTPUT RESULTS FOR CASE ONE
C
      WRITE (3,107) A,B
      WRITE (3,100) TT,NDATA
      WRITE (3,103) P,W,(Y(I),T(I),PRED(I),I=1,NDATA)
      P(3)=Y(NDATA)
C
C** PERFORM A GRID SEARCH TO OBTAIN INITIAL PARAMETER ESTIMATES FOR CASE TWO
C
      CALL GRIDS (30,30,FLIM,P,Y,NDATA,DIST,T,W,PRED)
      WRITE (3,107) P
      ES=20.
C
C
      )T,Y,MNN,ZZZ,001,5-,SE,E,P,DERP,3,ATADN( A20AV LLAC
C
      )2/3*3+ATADN(*4+3=MNN
      NNM=2+3*(NDATA+2*3/2)
C
C** PERFORM POWELL SEARCH FOR CASE TWO
C
      CALL VA02A (NDATA,2,PRED,P,E,ES,-20,200,ZZZ,NNM,Y,T)
      CALL DIST1 (PRED,NDATA,T,P)
      W=0.
      DO 302 I=1,NDATA
302  W=W+(Y(I)-PRED(I))**2.0
C
C** OUTPUT RESULTS FOR CASE TWO
C
      WRITE (3,100) TT,NDATA
      WRITE (3,103) P,W,(Y(I),T(I),PRED(I),I=1,NDATA)
      A=(P(1)+1.)/P(2)
      B=(P(1)+1.)/P(2)**2.
103  FORMAT (4E12.4/13(3F10.4/)/)
      WRITE (3,107) A,B
      GO TO 9
99   CALL EXIT
      END

```

```
// FOR MODSQ          DIST PARA BATCH WITH VAO2A
  SUBROUTINE MODSQ (FF,M,N,P,F,TT)
C
C** SUBROUTINE FOR POWELL REGRESSION OF BATCH DATA FOR THE GAMMA FUNCTION MODEL
C
  DIMENSION FF(1),P(3),F(1),TT(1)
  DO 1 I=1,M
    XXX=(P(2)/(P(2)+TT(I)))*(P(1)+1.)*(1.-P(3))+P(3)
1    F(I)=(FF(I)-XXX)
  RETURN
  END
```



// XEQ REGDL FX	
BATCH RD FEED COPPER 8/8/74 NO DEXTRIN TO RC	831 080811620
04007	
1000040603360283025402200203	
0000050010001500200030004000	
BATCH RD FEED NICKEL 8/8/74 NO DEXTRIN TO RC	841 080811620
04007	
1000066304850390035803140298	
0000050010001500200030004000	
BATCH RD FEED SULF 8/8/74 NO DEXTRIN TO RC	811 080811620
04007	
1000064305270363028401790148	
0000050010001500200030004000	
BATCH RD FEED GANGUE 8/8/74 NO DEX TO RC	821 080811620
04007	
1000098209680953094009200908	
0000050010001500200030004000	
MINERAL BATCH RD FEED INFINITE NWTST	811 010311625
04007	
1000055005120407037403600333	
00000500100015002000300040005000	
MINERAL BATCH RD FEED INFINITE NWTST	811 080311935
04007	
1000057405120401035102870260	
00000500100015002000300040005000	
MINERAL BATCH RD FEED INFINITE NWTST	81100140211805
04007	
1000057104940354030402360204	
0000050010001500200030004000	



```
// FOR INT
*IOCS(CARD)
C
C** USED TO INITIATE THE TWO PARAMETER GRID SEARCH PROGRAM FOR THE REGRESSION OF
C  BATCH DATA FOR THE TWO PHASE MODEL
C** ILC - TEST CODE
C** AA - LIMITS OF THE SEARCH AREA ON PULP TO FROTH RATE PARAMETER
C** BB - LIMITS OF THE SEARCH AREA ON THE FROTH TO PULP RATE PARAMETER
C
      EXTERNAL SEAR2
      COMMON DATA(13),TIME(13),PRED(13),ISA(50),ID(7),IS(30),ILC(7),AA(2
1),BB(2),F(13),NDATA,DB,DA,IWT(2),SG(13)
C**      ASK FOR DATA SPECIFICATIONS
C**      FITS TWO PARAMETER MODELS BY SEARCHING
C**      ALSO PLOTS SUM OF SQUARES SURFACE FOR THE SEARCH AREA
      READ (2,100) ILC,AA,BB,IWT
100      FORMAT (7I5,4F8.2,2I5)
      READ (2,101) (F(I),I=2,13)
101      FORMAT (13F6.4)
      F(1)=0.
      DA=(AA(2)-AA(1))/50.
      DB=(BB(2)-BB(1))/50.
      CALL LINK (SEAR2)
      END
```

```

// FOR SEAR2
*IOCS(1443 PRINTER)
*IOCS(CARD)
C
C** MAIN PROGRAM FOR TWO PARAMETER GRID SEARCH
C** TIME - BATCH FLOTATION TIME
C** DATA - OBSERVED SPECIES TAILINGS CONCENTRATION
C** PRED - FLOTATION MODELS BEST FIT OF THE TAILINGS CONCENTRATION
C** F - CONTAINS FROTH FLOW/FROTH VOLUME (RECIPROCAL MINS)
C** IS, ID - TEST DESCRIPTION
C** SAVA, SAVB - BEST FIT PARAMETERS
C** S - SUM OF SQUARED DEVIATIONS
C
      EXTERNAL INT
      DIMENSION P(11)
      COMMON DATA(13), TIME(13), PRED(13), ISA(50), ID(7), IS(30), ILC(7), AA(2
1), BB(2), F(13), NDATA, DB, DA, IWT(2), SG(13)
      DATA P/0.,.0005,.001,.005,.01,.05,.1,.5,1.,5.,1000./
      READ (2,110) IS, ID, NDATA
110  FORMAT (30A2,6X,3I1,I2,I4,I1,I4/2X,I3)
      READ (2,111) (DATA(I),I=1,NDATA)
111  FORMAT (20F4.3)
      READ (2,112) (TIME(I),I=1,NDATA)
112  FORMAT (20F4.2)
12   WRITE (3,101)
101  FORMAT (1H1)
      W=10.**10.
      DO 1 I=1,50
      A=AA(2)-DA*I
      DO 11 J=1,50
      B=BB(1)+DB*J
      ISA(J)=9
C
C** CALL TO TWO PHASE MODEL SUBROUTINE TO GENERATE PREDICTED SPECIES
C CONCENTRATIONS
C
      CALL MODSQ (A,B,S)
      DO 3 K=1,10
      MK=K+1
      XX=(P(MK)+P(K))/2.

```



```

      IF (ABS(S-XX)-(P(MK)-P(K))/2.) 4,4,3
4     ISA(J)=K-1
3     CONTINUE
      IF(W-S) 11,11,2
2     W=S
      SAVA=A
      SAVB=B
11    CONTINUE
1     WRITE (3,102) ISA
102   FORMAT (8X,50I2)
      CALL MODSQ (SAVA,SAVB,S)

```

C

C\*\* OUTPUT SECTION

C

```

      WRITE (3,103) IS,ID,SAVA,SAVB,S,AA,BB
103   FORMAT (/1X,30A2,7I5/' PARAMETERS = ',2E12.4/' SUM OF SQUARES = '
        ' ,E12.4/' SEARCH AREA = ',2E12.4,5X,2E12.4)
      WRITE (3,1000) (TIME(I),DATA(I),PRED(I),F(I),SG(I),I=1,NDATA)
1000  FORMAT (5E12.4)
7     CONTINUE
      CALL LINK (INT)
      END

```



```

// XEQ INT      FX
   8   1   1   0 1302   1 1915 0.   100.   0.   100.   1
.104 .0801 .0697 .0714 .0637 .0645
MINERAL BATCH RD FEED 100/5 NWTST      81100130211915
04007
1000076206320576053804770431
0000050010001500200030004000
   8   1   1   0 1402   1 1805 0.   100.   0.   100.   1   0
.279 .261 .320 .233 .334 .280
MINERAL BATCH RD FEED INFINITE NWTST    81100140211805
04007
1000057104940354030402360204
0000050010001500200030004000
   3   1   1   0 1402   1 1930 0.   100.   0.   100.   1   0
.534 .443 1.01 1.13
MINERAL BATCH RD CONC INFINITE NWTST    31100140211930
04005
10000220010400790069
00000500100015002000
   3   1   1   0 1102   1 2145 0.   100.   0.   100.   1
.0162 .0185 .0151 .0175 .0217 .0245
MINERAL BATCH RD CONC 100/5 NWTST      31100110212145
04007
1000077505760419031901960111
0000050010001500200030004000

```

```

// FOR GCONR                                GAMMA CONT REGRESSION
*IOCS(TYPEWRITER)
*IOCS(CARD)
C
C** PROGRAM TO ESTIMATE THE GAMMA FUNCTION MODEL RATE PARAMETERS FROM
C CONTINUOUS TEST DATA
C** T - MEAN CELL RETENTION TIMES * CUMULATIVE
C** Y - OBSERVED SPECIES CONCENTRATIONS
C** PRED - PREDICTED SPECIES CONCENTRATIONS
C** IS, ID - TEST IDENTIFICATION ARRAYS
C** P - BEST ESTIMATED RATE DISTRIBUTION PARAMETERS
C
      DIMENSION T(21),Y(21),PRED(21),ID(7),IS(30),P(4),E(2),ZZZ(100)
      DATA E/.01,.05/
98      CALL BREAD (T,Y,IS,ID,NDATA)
C
C** ESTIMATE FIRST PARAMETER VALUES FROM BATCH REGRESSION OF CONTINUOUS DATA
C
      W=10.**10.
      DEL=10.
      P(2)=-9.99
      DO 300 L=1,4
      DO 400 K=1,10
      P(2)=P(2)+DEL
      X=0.
      XX=0.
      DO 2 J=2,NDATA
      X=X+ALOG(Y(J))*ALOG(P(2)/(P(2)+T(J)))
2      XX=XX+(ALOG(P(2)/(P(2)+T(J))))**2.
      P(1)=X/XX-1.0
      CALL DIST1(PRED,NDATA,T,P)
      S=0.
      DO 3 J=1,NDATA
3      S=S+(Y(J)-PRED(J))**2.
      IF(S-W) 401,402,402
401  X1=W
      W=S
      A=P(1)
      B=P(2)
      GO TO 400

```



```
402 P(2)=P(2)-2.*DEL
    IF(P(2)) 404,405,405
404 P(2)=.01
405 DEL=DEL/5.
    W=X1
    GO TO 300
400 CONTINUE
300 CONTINUE
    P(1)=A
    P(2)=B
```

C

C\*\* INITIAL PARAMETER ESTIMATE FOR POWELL SEARCH COMPLETED

C

```
    ES=10.
    P(3)=(P(1)+1.)/P(2)
    P(4)=P(3)/P(2)
    NNM=2+3*(NDATA+2*3/2)
```

C

C\*\* START POWELL SEARCH

C

```
    CALL VAO2A (NDATA,2,PRED,P,E,ES,-5,200,ZZZ,NNM,Y,T)
```

C

C\*\* OUTPUT DATA AND MODEL FIT

C

```
    WRITE (1,100) IS,ID,P
100  FORMAT (//30A2,2X,7I5/4E16.6/)
    CALL MODSQ (Y,NDATA,100,P,PRED,T)
    XX=0.
    DO 1000 I=1,NDATA
    XX=XX+(Y(I)-PRED(I))*2.
    WRITE (1,101) Y(I),T(I),PRED(I),XX
1000 CONTINUE
101  FORMAT (4E16.6)
    GO TO 98
99   CALL EXIT
    END
```



```

// FOR MODSQ                      GAMMA CONT REGRESSION
SUBROUTINE MODSQ (Y, NDATA, N, P, PRED, T)
C
C** SUBROUTINE VARIATION FOR CONTINUOUS FLOTATION DATA * GENERATES THEORETICAL
C VALUES FOR USE BY VAO2A
C** USES SUBROUTINE KROPH TO GENERATE ROOTS AND WEIGHTS FOR A GENERALIZED GAUSS
C LAGUERRE QUADRATURE INTEGRATION OF THE FORM DESCRIBED IN PAGES 139-140
C
C** Y - OBSERVED NORMALIZED SPECIES CONCENTRATIONS
C** NDATA - NUMBER OF OBSERVATIONS
C** N - NUMBER OF VARIABLES I. E. 2 EXCEPT IN A SPECIAL CASE WHERE PRED ARRAY
C CONTAINS PREDICTED SPECIES CONCENTRATIONS (N=100)
C** P - PARAMETER VALUES OF THE RATE CONSTANT DISTRIBUTION, P(1) IS MEAN**2/
C VARIANCE -1, P(2) IS MEAN/VARIANCE
C** PRED - CONTAINS THE DIFFERENCE BETWEEN OBSERVED AND PREDICTED SPECIES
C CONCENTRATIONS EXCEPT AS ABOVE UNDER N
C** T - CONTAINS MEAN FLOTATION TIMES CORRESPONDING TO Y
C
      DIMENSION Y(1), PRED(1), T(2), P(4), RTZ(15), W(15)
      X=P(1)
      D1=RCGAM(X+1.000)
      CALL KROPH(X, D1, RTZ, W, 15)
      DO 300 I=1, NDATA
        ZED=I-1
        PRED(I)=0.
        DO 301 K=1, 15
301      PRED(I)=PRED(I)+W(K)/D1/(1.+RTZ(K)*T(2)/P(2))**ZED
        IF(N=100) 290, 300, 290
290      PRED(I)=Y(I)-PRED(I)
300      CONTINUE
      RETURN
      END

```

// XEQ GCONR FX  
MINERAL PLANT RUN 34 CLEANER 22/6/72 41134220621215  
04021  
10000889075606860641061305980577055005390519050904960486047104590449043604260414  
0403  
0 64 129 193 258 323 387 452 517 581 646 711 775 840 905 9691034109911631228  
1293  
MINERAL PLANT CL FEED 37 41137130722115  
04021101000400700010  
10000850067105930514048204330382033003100283027302600247022902140200019001780167  
0155  
0 72 144 217 290 363 435 508 581 654 726 799 872 945101710901163123613081381  
1454  
PLANT RUN 40 9/8/72 MINERAL 41140090822000  
04021  
10000781069806240588055005180417036903470314030302860251022101930171015301400126  
0114  
0 79 159 239 319 399 479 558 638 719 799 878 9581038111811981278135714381518  
1598  
GANGUE PLANT TEST CL FEED 34 42134220621215  
04021102001103000010  
10000990097409620954094509410933092309160907090008920884087508670860084908380825  
0813  
0 64 129 193 258 323 387 452 517 581 646 711 775 840 905 9691034109911631228  
1293







```

// FOR VA02A                                POWELL SEARCH TECHNIQUES
      SUBROUTINE  VA02A(M,N,F,X,E,STEP,LIST,MXITN,W,NNM,F9,TT)
C
C** SUBROUTINE TO IMPLEMENT POWELL SEARCH TECHNIQUES.  USES VA01A AND USER
C   SUPPLIED MODSQ.
C** M IS THE NUMBER OF DATA POINTS
C** N - IS THE NUMBER OF PARAMETERS
C** F - IS AN ARRAY OF M LENGTH CONTAINING THE SQUARE OF THE ERROR AT EACH POINT
C** X - IS AN ARRAY OF N LENGTH CONTAINING PARAMETER VALUES
C** E - IS AN ARRAY OF PARAMETER TOLERANCES AS EXIT CRITERION
C** STEP - IS A FACTOR WHICH WHEN MULTIPLIED BY E GIVES THE MAXIMUM SINGLE MOVE
C   MADE IN THE X PARAMETER VALUE
C** LIST - IF ZERO NO PRINTING IS DONE.  IF LIST GREATER THAN ZERO, DETAILS
C   OF THE SEARCH ARE PRINTED EVERY LIST ITERATIONS.  IF LIST IS LESS THAN ZERO
C   PRINTING IS THE SAME AS CASE TWO EXCEPT FUNCTIONAL VALUES ARE NOT PRINTED.
C** MXITN - MAXIMUM NUMBER OF ITERATIONS BEFORE EXIT
C** W - AN ARRAY OF AT LEAST NNM REAL WORDS
C** NNM -  $2+3*(M+2*3/2)$ 
C** F9 - AN ARRAY OF M LENGTH CONTAINING OBSERVED DATA POINTS * Y OR ORDINATES
C** TT - AN ARRAY OF M LENGTH HOLDING X OR ABCISSA VALUES FOR F9
C
      DIMENSION X(1),E(1),W(1),F(1),F9(1),TT(1)
      LUND=LUND
      MANDN=M+N
      KST=N+MANDN
      NPLUS=N+1
      KINV=NPLUS*(MANDN+1)
      KSTO=KINV-MANDN-1
C
C** USER SUPPLIED
C
      CALL MODSQ (F9,M,N,X,F,TT)
      NN=N+N
      K=NN
      DO1 I=1,M
      K=K+1
      W(K)=F(I)
1  CONTINUE
      IINV=2
      K=KST

```

```

      I=1
      2 X(I)=X(I)+E(I)
C
C** USER SUPPLIED
C
      CALL MODSQ (F9,M,N,X,F,TT)
      X(I)=X(I)-E(I)
      DO 3 J=1,N
      K=K+1
      W(K)=0.
      W(J)=0.
      3 CONTINUE
      SUM=0.
      KK=NN
      DO 4 J=1,M
      KK=KK+1
      F(J)=F(J)-W(KK)
      SUM=SUM+F(J)*F(J)
      4 CONTINUE
      IF (SUM) 5,5,6
5      CONTINUE
C
      WRITE (1,7) I
      7 FORMAT (5X,8HVA02A E(,I3,20H) UNREASONABLY SMALL)
      DO 8 J=1,M
      NN=NN+1
      F(J)=W(NN)
      8 CONTINUE
      GO TO 10
      6 SUM=1./SQRT(SUM)
      J=K-N+1
      W(J)=E(I)*SUM
      DO 9 J=1,M
      K=K+1
      W(K)=F(J)*SUM
      KK=NN+J
      DO 11 II=1,I
      KK=KK+MANDN
      W(II)=W(II)+W(KK)*W(K)
      11 CONTINUE

```

I )7,3( ETIRW

C



```

9  CONTINUE
   ILESS=I-1
   IGAMA=N+I-1
   ICINV=N-ILESS
   ICINP=ICINV+1
   IF (ILESS) 13,13,14
13  W(KINV)=1.
   GO TO 15
14  B=1.
   DO 16 J=NPLUS,IGAMA
     W(J)=0.
16  CONTINUE
   KK=KINV
   DO 17 II=1,ILESS
     IIP=II+N
     W(IIP)=W(IIP)+W(KK)*W(II)
     JL=II+1
     IF (JL-ILESS) 18,18,19
18  DO 20 JJ=JL,ILESS
     KK=KK+1
     JJP=JJ+N
     W(IIP)=W(IIP)+W(KK)*W(JJ)
     W(JJP)=W(JJP)+W(KK)*W(II)
20  CONTINUE
19  B=B-W(II)*W(IIP)
   KK=KK+ICINP
17  CONTINUE
   B=1./B
   DO 21 II=NPLUS,IGAMA
     BB=-B*W(II)
   DO 22 JJ=II,IGAMA
     W(KK)=W(KK)-BB*W(JJ)
   KK=KK+1
22  CONTINUE
   W(KK)=BB
   KK=KK+ICINV
21  CONTINUE
   W(KK)=B
15  GO TO (27,24),IINV
24  I=I+1

```



```

      IF (I-N) 2,2,25
25  IINV=1
      FF=0.
      KL=NN
      DO 26 I=1,M
      KL=KL+1
      F(I)=W(KL)
      FF=FF+F(I)*F(I)
26  CONTINUE
      ICONT=1
      ISS=1
      MC=N+1
      IPP=IABS(LIST)*(IABS(LIST)-1)
      ITC=0
      IPS=1
      IPC=0
27  IPC=IPC-IABS(LIST)
      IF (IPC) 28,29,29
28  CONTINUE
C                                     FF,CM,CTI )03,3( ETIRW
      WRITE (1,30) ITC,MC,FF
30  FORMAT (//5X,9HITERATION,I4,I9,16H CALLS OF CALCUN,5X,2HF=,E24.14)
C                                     )N,1=I,)I(X( )13,3( ETIRW
      WRITE (1,31) (X(I),I=1,N)
31  FORMAT (5X,9H VARIABLES,/(5E24.14))
      IF(LIST)1000,1001,1001
1001 CONTINUE
C                                     )M,1=I,)I(F( )23,3( ETIRW
      WRITE (1,32) (F(I),I=1,M)
32  FORMAT (5X,9HFUNCTIONS,/(5E24.14))
1000 IPC=IPP
      GO TO (29,33),IPS
29  GO TO (34,35),ICONT
35  IF(CHANJ-1.)10,10,36
10  IF(LIST)1003,33,37
1003 CONTINUE
C                                     )2001,3( ETIRW
      WRITE (1,1002)
1002 FORMAT(//5X,31HVA02A FINAL VALUES OF VARIABLES)
      GO TO 1004

```

```

37  CONTINUE
C                                     )83,3( ETIRW
    WRITE (1,38)
38  FORMAT (//5X,45HVA02A FINAL VALUES OF FUNCTIONS AND VARIABLES)
1004 IPS=2
    GO TO 28
33  RETURN
36  ICONT=1
34  ITC=ITC+1
    K=N
    KK=KST
    DO 39 I=1,N
    K=K+1
    W(K)=0.
    KK=KK+N
    W(I)=0.
    DO 40 J=1,M
    KK=KK+1
    W(I)=W(I)+W(KK)*F(J)
40  CONTINUE
39  CONTINUE
    DM=0.
    K=KINV
    DO 41 II=1,N
    IIP=II+N
    W(IIP)=W(IIP)+W(K)*W(II)
    JL=II+1
    IF (JL-N) 42,42,43
42  DO 44 JJ=JL,N
    JJP=JJ+N
    K=K+1
    W(IIP)=W(IIP)+W(K)*W(JJ)
    W(JJP)=W(JJP)+W(K)*W(II)
44  CONTINUE
    K=K+1
43  IF (DM-ABS(W(II)*W(IIP))) 45,41,41
45  DM=ABS(W(II)*W(IIP))
    KL=II
41  CONTINUE
    II=N+MANDN*KL

```



```

      CHANJ=0.
      DO 46 I=1,N
      JL=N+I
      W(I)=0.
      DO 47 J=NPLUS,NN
      JL=JL+MANDN
      W(I)=W(I)+W(J)*W(JL)
47  CONTINUE
      II=II+1
      W(II)=W(JL)
      W(JL)=X(I)
      IF (ABS(E(I)*CHANJ)-ABS(W(I)))48,48,46
48  CHANJ=ABS(W(I)/E(I))
46  CONTINUE
      DO 49 I=1,M
      II=II+1
      JL=JL+1
      W(II)=W(JL)
      W(JL)=F(I)
49  CONTINUE
      FC=FF
      ACC=0.1/CHANJ
      IT=3
      XC=0.
      XL=0.
      IS=3
      XSTEP=-AMIN1(0.5,STEP/CHANJ)
      IF (CHANJ-1.)50,50,51
50  ICONT=2
51  CALL VD01A(IT,XC,FC,6,ACC,0.1,XSTEP)
      GO TO (52,53,53,53),IT
52  MC=MC+1
      IF (MC-MXITN)54,54,55
55  CONTINUE
      C
      WRITE (1,56) MXITN
56  FORMAT(5X,5HVA02A,I6,17H  CALLS OF CALFUN)
      ISS=2
      GO TO 53
54  XL=XC-XL

```

ATTACHED

ATTACHED

NTIXM )65,3( ETIRW

C



```

        DO 57 J=1,N
        X(J)=X(J)+XL*W(J)
57 CONTINUE
        XL=XC
C
C** USER SUPPLIED
C
        CALL MODSQ (F9,M,N,X,F,TT)
        FC=0.
        DO 58 J=1,M
        FC=FC+F(J)*F(J)
58 CONTINUE
        GO TO (59,59,60),IS
60 K=N
        IF (FC-FF) 61,51,62
61 IS=2
        FMIN=FC
        FSEC=FF
        GO TO 63
62 IS=1
        FMIN=FF
        FSEC=FC
        GO TO 63
59 IF (FC-FSEC) 64,51,51
64 K=KST0
        GO TO (75,74),IS
75 K=N
74 IF (FC-FMIN) 65,51,66
66 FSEC=FC
        GO TO 63
65 IS=3-IS
        FSEC=FMIN
        FMIN=FC
63 DO 67 J=1,N
        K=K+1
        W(K)=X(J)
67 CONTINUE
        DO 68 J=1,M
        K=K+1
        W(K)=F(J)

```

```
68 CONTINUE
GO TO 51
53 K=KSTO
KK=N
GO TO (69,70,69),IS
70 K=N
KK=KSTO
69 SUM=0.
DM=0.
JJ=KSTO
DO 71 J=1,N
K=K+1
KK=KK+1
JJ=JJ+1
X(J)=W(K)
W(JJ)=W(K)-W(KK)
71 CONTINUE
DO 72 J=1,M
K=K+1
KK=KK+1
JJ=JJ+1
F(J)=W(K)
W(JJ)=W(K)-W(KK)
SUM=SUM+W(JJ)*W(JJ)
DM=DM+F(J)*W(JJ)
72 CONTINUE
GOTO(73,10),ISS
73 J=KINV
KK=NPLUS-KL
DO 76 I=1,KL
K=J+KL-I
J=K+KK
W(I)=W(K)
W(K)=W(J-1)
76 CONTINUE
IF (KL-N) 77,78,78
77 KL=KL+1
JJ=K
DO 79 I=KL,N
K=K+1
```



```

      J=J+NPLUS-1
      W(I)=W(K)
      W(K)=W(J-1)
79  CONTINUE
      W(JJ)=W(K)
      B=1./W(KL-1)
      W(KL-1)=W(N)
      GO TO 88
78  B=1./W(N)
88  K=KINV
      DO 80 I=1,ILESS
      BB=B*W(I)
      DO 81 J=I,ILESS
      W(K)=W(K)-BB*W(J)
      K=K+1
81  CONTINUE
      K=K+1
80  CONTINUE
      IF (FMIN-FF) 82,83,83
83  CHANJ=0.
      GO TO 84
82  FF=FMIN
      CHANJ=ABS(XC)*CHANJ
84  XL=-DM/FMIN
      SUM=1./SQRT(SUM+DM*XL)
      K=KSTO
      DO 85 I=1,N
      K=K+1
      W(K)=SUM*W(K)
      W(I)=0.
85  CONTINUE
      DO 86 I=1,M
      K=K+1
      W(K)=SUM*(W(K)+XL*F(I))
      KK=NN+I
      DO 87 J=1,N
      KK=KK+MANDN
      W(J)=W(J)+W(KK)*W(K)
87  CONTINUE
86  CONTINUE

```



GO TO 14  
END

```
// FOR VD01A      ( ITEST,X,F,MXITN,ABSAC,RELAC,XSTEP)
SUBROUTINE VD01A(ITEST,X,F,MXITN,ABSAC,RELAC,XSTEP)
```

```
C**
```

```
C** USED BY VA02A * NOT USER SPECIFIC
```

```

      GO TO (1,2,2),ITEST
2  IS=6-ITEST
   ITEST=1
   IINC=1
   XINC=XSTEP+XSTEP
   MC=IS-3
   IF (MC)4,4,15
3  MC=MC+1
   IF (MXITN-MC)12,15,15
12 ITEST=4
43 X=DB
   F=FB
   IF (F-B-FC)15,15,44
44 X=DC
   F=FC
15 RETURN
   1 GO TO (5,6,7,8),IS
   8 IS=3
   4 DC=X
   FC=F
   X=X+XSTEP
   GO TO 3
   7 IF (FC-F)9,10,11
10 X=X+XINC
   XINC=XINC+XINC
   GO TO 3
   9 DB=X
   FB=F
   XINC=-XINC
   GO TO 13
11 DB=DC
   FB=FC
   DC=X
   FC=F
13 X=DC+DC-DB
```



```

      IS=2
      GO TO 3
6    DA=DB
      DB=DC
      FA=FB
      FB=FC
32   DC=X
      FC=F
      GO TO 14
5    IF (FB-FC) 16,17,17
17   IF (F-FB) 18,32,32
18   FA=FB
      DA=DB
19   FB=F
      DB=X
      GO TO 14
16   IF (FA-FC) 21,21,20
20   XINC=FA
      FA=FC
      FC=XINC
      XINC=DA
      DA=DC
      DC=XINC
21   XINC=DC
      IF ((D-DB)*(D-DC)) 32,22,22
22   IF (F-FA) 23,24,24
23   FC=FB
      DC=DB
      GO TO 19
24   FA=F
      DA=X
14   IF (FB-FC) 25,25,29
25   IINC=2
      XINC=DC
      IF (FB-FC) 29,45,29
29   D=(FA-FB)/(DA-DB)-(FA-FC)/(DA-DC)
      IF (D*(DB-DC)) 33,33,37
37   D=0.5*(DB+DC-(FB-FC)/D)
      IF (ABS(D-X) - ABS(ABSAC)) 34,34,35
35   IF (ABS(D-X) - ABS(D*RFIAC)) 34,34,36

```



```
34 ITEST=2
   GO TO 43
36 IS=1
   X=D
   IF((DA-DC)*(DC-D))3,26,38
38 IS=2
   GO TO (39,40),IINC
39 IF(ABS(XINC)-ABS(DC-D))41,3,3
33 IS=2
   GO TO (41,42),IINC
41 X=DC
   GO TO 10
40 IF(ABS(XINC-X)-ABS(X-DC))42,42,3
42 X=0.5*(XINC+DC)
   IF((XINC-X)*(X-DC))26,26,3
45 X=0.5*(DB+DC)
   IF((DB-X)*(X-DC))26,26,3
26 ITEST=3
   GO TO 43
   END
```

```
// FOR AMIN1      F(A,B)
      FUNCTION AMIN1(A,B)
C
C** COMPARES NUMBERS A AND B, MINIMUM NUMBER IS PASSED TO LEFT HAND SIDE
C
      IF(A-B)20,20,10
10    AMIN1=B
      GOTO30
20    AMIN1=A
30    RETURN
      END
```

```
// FOR DIST
      SUBROUTINE DIST (PRED,N,T,P)
C
C** SUBROUTINE TO GIVE GAMMA FUNCTION MODEL BATCH CONCENTRATIONS FROM PARAMETERS
C   AND TIME ARRAY * USED BY REGDL
C
      DIMENSION PRED(1),T(1),P(3)
      DO 1 I=1,N
        PRED(I)=(P(2)/(P(2)+T(I)))**(P(1)+1.)
        PRED(I)=PRED(I)*(1.-P(3))-P(3)
      1 CONTINUE
      RETURN
      END
```

NF\_ 2



```
// FOR DIST1
      SUBROUTINE DIST1(PRED,N,T,P)
C
C** SUBROUTINE TO CALCULATE THE GAMMA FUNCTION MODEL BATCH CURVE GIVEN FLOAT
C   TIMES AND PARAMETERS
C** PRED - AN ARRAY OF LENGTH N CONTAINING CALCULATED NORMALIZED CONCENTRATIONS
C** N - NUMBER OF FLOAT TIMES TO BE EVALUATED
C** T - AN ARRAY OF LENGTH N CONTAINING THE BATCH FLOTATION TIMES
C** P - AN ARRAY OF RATE DISTRIBUTION PARAMETERS
C   P(1) IS MEAN**2/VARIANCE-1
C   P(2) IS MEAN/VARIANCE
C   P(3) IS THE NORMALIZED FRACTION OF NONFLOATING MATERIAL
C
      DIMENSION PRED(1),T(1),P(3)
      DO 1 I=1,N
        PRED(I)=(P(2)/(P(2)+T(I)))**(P(1)+1.)
        PRED(I)=PRED(I)*(1.-P(3))+P(3)
1      CONTINUE
      RETURN
      END
```

NFL C

```

// FOR BCONT                ALPHA CN CONTINUOUS PREDICTS
*IOCS(CARD)
*IOCS(1443 PRINTER)
C
C** PROGRAM TO PREDICT CONTINUOUS BANK FLOTATION GIVEN BATCH TEST PARAMETERS FOR
C THE RATE PLUS NONFLOATING MODEL AND THE SINGLE RATE MODEL
C** T - CONTAINS OBSERVED MEAN RETENTION TIMES FOR THE FLOTATION BANK
C** Y - CONTAINS ACTUAL OBSERVED SPECIES NORMALIZED CONCENTRATIONS FOR THE
C FLOTATION BANK
C** P - CONTAINS PREDICTED SPECIES CONCENTRATIONS FOR THE FLOTATION BANK
C** A - CONTAINS FIRST ORDER RATE CONSTANT AND THE NONFLOATING FRACTION OF THE
C SPECIES
C** IS, ID - TEST IDENTIFICATION (CONTINUOUS TEST)
C** S - PROGRESSIVE SUM OF SQUARES OF THE DEVIATIONS BETWEEN OBSERVED (Y) AND
C PREDICTED (P)
C
      DIMENSION IS(30),ID(7)
      DIMENSION T(21),Y(21),P(21),A(2)
2      CONTINUE
      CALL RDCRD (1,NNV,T)
      CALL RDCRD (2,NV,A)
      CALL BREAD (T,Y,IS,ID,NDATA)
100     FORMAT (30A2/7I5)
      WRITE (3,100) IS,ID
      WRITE (3,101) A
      S=0.
      DO 1 I=1,NDATA
      ZED=I-1
      P(I)=(1.-A(2))/(1.+A(1)*T(2))**ZED+A(2)
      S=S+(Y(I)-P(I))**2.0
1      WRITE (3,101) T(I),Y(I),P(I),S
101     FORMAT (4E12.4)
      GO TO 2
99     CALL EXIT
      END

```

DUMMY



// XEQ BCONT FX

-.04173 8.868

.1052 .207

ROUGHER A3 13/4/73 1435 MINERAL

811 130421435

04021

10000912076506160578056705580538051304430417040904010388038203760360035103400331

0321

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

-.9256 12.65

.04409 .881

ROUGHER A3 13/4/73 1435 GANGUE

821 130421435

04021

10000998099309870984098309820979097509710969096809670966096409640962096109580957

0956

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

-.5519 .3572

.3358 .154

ROUGHER BANK COPPER 13/4/72 1430

831 130421430

04021

10000695047903730354034603420331031702940282027702740268026202590250024502360232

0226

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

-.4263 3.593

.1385 .271

ROUGHER BANK NICKEL 13/4/72 1430

841 130421430

04021

10000934076306280597058205730557053705160506050305000495049204910486048204780476

0472

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

-.3763 3.654

.1341 .236

ROUGHER A3 23/4/73 1200 MINERAL

811 230421200

04021



```

// FOR GCONT          GAMMA CONTINUOUS PREDICTS
*IOCS(TYPEWRITER)
*IOCS(CARD)
C
C** PROGRAM TO READ GAMMA FUNCTION MODEL BATCH PARAMETERS AND CONTINUOUS TEST
C DATA. BATCH PARAMETERS ARE USED TO PREDICT CONTINUOUS OPERATION.
C SEE PAGES 71 AND 138
C** T - CONTAINS MEAN RETENTION TIMES FOR CONTINUOUS TEST BANK OF CELLS
C** Y - CONTAINS OBSERVED SPECIES CONCENTRATIONS AT TIME T.
C** IS, ID - CONTAINS CONTINUOUS TEST DESCRIPTION AND CODE
C** P - CONTAINS PREDICTED SPECIES CONCENTRATIONS
C** A - CONTAINS BATCH TEST REGRESSED PARAMETERS
C** RTZ, W - CONTAINS ROOTS AND WEIGHTS FOR GAUSS-LAGUERRE QUADRATURE INTEGRATION
C** SS - CONTAINS PROGRESSIVE SUM OF SQUARES OF THE DEVIATION BETWEEN Y AND P
C
      DIMENSION T(21),Y(21),ID(7),P(21),IS(30)
      DIMENSION RTZ(15),W(15)
      DIMENSION A(4)
      IOUT=3
98      READ (2,200) A
      READ (2,200) A
      A1=A(1)
      D1=RCGAM(A(1)+1.)
      CALL KROPH (A1,D1,RTZ,W,15)
      QUADD
200     FORMAT (4F12.5)
      CALL BREAD (T,Y,IS,ID,NDATA)
      DO 300 I=1,21
      ZED=I-1
      P(I)=0.
      DO 301 K=1,15
301     P(I)=P(I)+W(K)/D1/(1.+RTZ(K)*T(2)/A(2))**ZED
C          DEZ**))2(A/2(T*)K(ZTR+.1(/ID/))1(A**K(ZTR*))<(W+I(P=)I(P
103
300     CONTINUE
      A(3)=(A(1)+1.)/A(2)
      A(4)=A(3)/A(2)
99      WRITE (IOUT,100) IS,ID,A
      SS=0.
      DO 1 I=1,NDATA
      SS=SS+(Y(I)-P(I))**2.0
1      WRITE (IOUT,102) T(I),Y(I),P(I),SS

```

```
      WRITE (IOUT,103) (RTZ(I),W(I),I=1,15)
103  FORMAT (/ ' ROOTS AND WEIGHTS USED' /15(2E20.9/))
      GO TO 98
100  FORMAT (/30A2/7I5/4E12.4//)
102  FORMAT (4E12.4)
101  CALL EXIT
      END
```



// XEQ GCONT FX

.1052 .207

-.04173 8.868

ROUGHER A3 13/4/73 1435 MINERAL

811 130421435

04021

10000912076506160578056705580538051304430417040904010388038203760360035103400331

0321

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

.04409 .881

-.9256 12.65

ROUGHER A3 13/4/73 1435 GANGUE

821 130421435

04021

10000998099309870984098309820979097509710969096809670966096409640962096109580957

0956

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

.3358 .154

-.5519 .3572

ROUGHER BANK COPPER 13/4/72 1430

831 130421430

04021

10000695047903730354034603420331031702940282027702740268026202590250024502360232

0226

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

.1385 .271

-.4263 3.593

ROUGHER BANK NICKEL 13/4/72 1430

841 130421430

04021

10000934076306280597058205730557053705160506050305000495049204910486048204780476

0472

0 68 137 207 276 345 414 483 552 621 690 759 829 898 96710361105117412431312

1381

.1341 .236

-.3763 3.654



```

// FOR CLEAN          ROUGHER BATCH TO CLEANER BATCH PREDICTIONS
*IOCS(TYPEWRITER)
*IOCS(CARD)
C
C** CALCULATION OF ROUGHER CONCENTRATE RATE CONSTANT DISTRIBUTION (GAMMA MODEL)
C FROM THE RATE CONSTANT DISTRIBUTION OF THE FEED TO THE ROUGHER BANK AS
C DEFINED BY A BATCH FLOTATION TEST. SEE PAGE 136
C** IS, ID - TEST IDENTIFICATION (ROUGHER CONCENTRATE FLOAT)
C** A - CONTAINS ROUGHER BATCH PARAMETERS AS REGRESSED
C** AA, BB, BARK, SIGMA - CONTAIN CALCULATED ROUGHER CONCENTRATE PARAMETER
C** Y - CONTAINS ACTUAL BATCH TEST DATA ON ROUGHER CONCENTRATE - NORMALIZED
C SPECIES CONCENTRATIONS
C** T - CONTAINS ROUGHER CONCENTRATE BATCH FLOTATION TIMES
C** P - CONTAINS SPECIES CONCENTRATIONS FOR THE ROUGHER CONCENTRATE BATCH FLOAT
C WHICH ARE PREDICTED THROUGH THE REGRESSED ROUGHER FEED BATCH PARAMETERS
C** SUMSQ - SUM OF THE SQUARES OF THE DEVIATIONS BETWEEN Y AND P
C
      DIMENSION RTZ(25), W(25), Y(7), T(7), A(5), P(7), IS(30), ID(7), RA(3)      C
      DIMENSION F(25)                                                            C
200  FORMAT (F12.5, F13.7, F11.5, F12.5, F12.2)
201  FORMAT (30A2, 6X, 3I1, I2, I4, I1, I4/2X, I3)
202  FORMAT (20F4.3)
203  FORMAT (20F4.2)
C
C** READ IN ROUGHER FEED BATCH PARAMETERS
C
98   READ (2, 200) A
      NX=20
      READ (2, 200) A
      READ (2, 200) A
      A(3)=A(1)
      A(4)=A(2)
      A(1)=A(3)**2./A(4)-1.000
      A(2)=A(3)/A(4)
      IF(A(1)+1.) 99, 99, 97
97   READ (2, 201) IS, ID, NDATA
      READ (2, 202) (Y(I), I=1, NDATA)
      READ (2, 203) (T(I), I=1, NDATA)
C
C** CALCULATE MEAN RETENTION TIME IN THE CELLS OF THE CONTINUOUS ROUGHER BANK

```

```

C      TT=4.*522./A(5)
      ALF=A(1)
      ALF=A(1)+1.0000
      DO 500 I=1,3
      RA(I)=0.
      GO TO 1
NEW KRPC

C      2,1,1 ).8-FLA(FI
NEW KRPC
2      DO 3 K=1,15
      F(K)=0.
      DO 3 L=1,20
      ZED=L
3      F(K)=F(K)+Z(K)**ALF*TT/(1.0+Z(K)*TT/A(2))**ZED
      DO 4 K=1,15
4      RA(I)=RA(I)+WZ(K)*F(K)
      GO TO 6
1      D1=RCGAM(ALF+1.)
      CALL KROPH (ALF,D1,RTZ,W,NX)
      DO 306 K=1,NX
      F(K)=0.
      DO 306 L=1,20
      ZED=L
C      DEZ**))2(A/TT*)K(ZTR+.1(/TT*)K(ZTR+)K(F=)K(F 603
306      F(K)=F(K)+      TT/(1.+RTZ(K)*TT/A(2))**ZED      NEW KRPC
      DO 301 K=1,NX
301      RA(I)=RA(I)+W(K)*F(K)
6      ALF=ALF+1.
500      CONTINUE
      RA(2)=RA(2)/A(2)
      RA(3)=RA(3)/A(2)**2.
      BARK=RA(2)/RA(1)
      SIGMA=(RA(3)-2.*BARK*RA(2)+BARK**2.*RA(1))/RA(1)
      AA=BARK**2./SIGMA-1.00000
      BB=BARK/SIGMA
      DO 501 I=1,NDATA
501      P(I)=(BB/(BB+T(I)))*(AA+1.0)
C
C** OUTPUT SECTION
C      WRITE (3,100) IS,ID,A,      AA,BB,BARK,SIGMA

```



```
      WRITE (3,101) (RA(I),I=1,3)
100  FORMAT (//30A2,5X,3I1,I2,I4,I1,I4/' ROUGHER ',5E20.9/' ROUGHER CON
      C ',4E20.9/)
      SUMSQ=0.
      DO 1001 I=1,NDATA
      SUMSQ=SUMSQ+(Y(I)-P(I))**2.
1001 WRITE (3,101) Y(I),T(I),P(I),SUMSQ
101  FORMAT (4E13.5)
      GO TO 98
99   CALL EXIT
      END
```



// XEQ CLEAN FX

.08672

.08825 .004221

3570.

.09586 .0004376

3570.

BATCH RO CONC NO DEX 8/8/74 SULF

311 080811645

04005

10000317019801480115

0000050010001500200030004000

.02779

.00409 .0001601

3570.

.05578 .000185

3570.

BATCH RO CONC NO DEX 8/8/74 GANGUE

321 080811645

04005

10000746065306060564

0000050010001500200030004000

.2722

.8467 2.065

3570.

.4336 .1347

3570.

BATCH RO CONC NO DEX 8/8/74 COPPER

331 080811620

04005

10000140009200730060

0000050010001500200030004000

.1318

.1486 .04277

3570.

.1332 .0003526

3570.

BATCH RO CONC NO DEX 8/8/74 NICKEL

341 080811620

04005

10000247014401130093

0000050010001500200030004000

.1968

.2468 .02787

3840.

.4024 .5815

3840.

MINERAL BATCH RO CONC INFINITE NWTST

311 010311800

04005

10000247016101380126

00000500100015002000300040005000

```

// FOR RDCRF          RD TO CL FOR CF,CNF,ALPHA MODEL
*IDCS(CARD)
*IDCS(TYPEWRITER,KEYBOARD)
C
C** PROGRAM TO PREDICT ROUGHER CONCENTRATE FLOTATION FROM RATE PLUS NONFLOATING
C  PARAMETERS REGRESSED FROM A BATCH TEST ON THE ROUGHER FEED
C** IS,ID - TEST IDENTIFICATION (ROUGHER CONCENTRATE BATCH)
C** Y - OBSERVED NORMALIZED CONCENTRATIONS
C** T - OBSERVED FLOAT TIMES
C** P - PREDICTED CONCENTRATIONS
C** ALPHA - ROUGHER FEED FIRST ORDER RATE CONSTANT
C** S - SUM OF SQUARED DEVIATIONS
C
      DIMENSION Y(21),T(21),P(21),ID(7),IS(30)
99      CONTINUE
      CALL RDCRD (1,NV,T)
      CALL RDCRD (1,NV,T)
      CALL RDCRD (1,NV,ALPHA)
      CALL BREAD (T,Y,IS,ID,NDATA)
      WRITE (3,2000) IS,ID,ALPHA
2000    FORMAT (30A2/7I5,F10.4)
      S=0.
      DO 1 I=1,NDATA
      P(I)=EXP(-ALPHA*T(I))
      S=S+(Y(I)-P(I))**2.0
1      WRITE (3,101) Y(I),T(I),P(I),S
101    FORMAT (4E12.4)
      GO TO 99
98      CALL EXIT
      END

```



// XEQ ROCLF FX

.08825 .004221

3570.

.09586 .0004376

3570.

.08672

BATCH RO CONC NO DEX 8/8/74 SULF

311 080811645

04005

10000317019801480115

0000050010001500200030004000

.00409 .0001601

3570.

.05578 .000185

3570.

.02779

BATCH RO CONC NO DEX 8/8/74 GANGUE

321 080811645

04005

10000746065306060564

0000050010001500200030004000

.8467 2.065

3570.

.4336 .1347

3570.

.2722

BATCH RO CONC NO DEX 8/8/74 COPPER

331 080811620

04005

10000140009200730060

0000050010001500200030004000

.1486 .04277

3570.

.1332 .0003526

3570.

.1318

BATCH RO CONC NO DEX 8/8/74 NICKEL

341 080811620

04005

10000247014401130093

0000050010001500200030004000

.2468 .02787

3840.

.4024 .5815

3840.

.1968

MINERAL BATCH RO CONC INFINITE NWTST

311 010311800

04005

10000247016101380126

00000500100015002000300040005000



```

// FOR ITIT                      H-R PRED FROM A,B
*IOCS(TYPEWRITER)
C
C** PROGRAM TO PREDICT BATCH FLOTATION THROUGH THE TWO PHASE MODEL GIVEN
C      Q - FROTH FLOW / FROTH VOLUME
C      T - FLOAT TIMES
C      A,B - FIRST ORDER FLOTATION AND FROTH RETURN RATE PARAMETERS
C** PRED - CONTAINS PREDICTED SPECIES CONCENTRATIONS AT TIMES T
C
      COMMON Y(13),T(13),PRED(13),ISA(50),ID(7),IS(30),ILC(7),AA(2),BB(2
      ),Q(13),NDATA,DB,DA,IWT(2),SG(13)
23  CONTINUE
C
C** TYPEWRITER FREE FORMAT INPUT TO ARRAY Y
C
      CALL LARRY (13,NV,Y)
      IWT(1)=Y(1)
      IWT(2)=Y(2)
      A=Y(3)
      B=Y(4)
      NDATA=Y(5)
      Y(NDATA)=Y(6)
      CALL LARRY (13,NV,Q)
      CALL LARRY (13,NV,T)
      CALL MODSQ (A,B,S)
      WRITE (1,100) A,B,S
      WRITE (1,100) (T(I),PRED(I),Q(I),I=1,NDATA)
100  FORMAT (3E12.4)
      GO TO 23
99  CALL EXIT
      END

```

```

// FOR PRFWO
*IOCS(CARD)
C
C** DATA INPUT PROGRAM FOR SIMULATION OF TESTED CIRCUITS BY THE GAMMA FUNCTION
C  MODEL.  SEE PAGES 121-127
C** P - CONTAINS SPECIES PARAMETERS OF THE FRESH FEED AS ESTIMATED FROM OTHER
C  TESTING.
C** FG - CONTAINS SPECIES FRACTION IN FRESH FEED
C** FV - CONTAINS FRESH FEED VOLUME FLOW RATE (L/MIN)
C
      EXTERNAL PRFW1
      DIMENSION A(2),B(2),FG(16,2),TN(16,2),GA(16),R(16,2),P(3,2),FV(16)
      COMMON/INSKEL/MMM,A,B,FG,TN,GA,R,P,FV
      MMM=2
      MMM=16
      MMM=8
      MMM=3
      MMM=1
      DO 1 L=1,MMM
      READ (2,100) FG(L,1),FV(L)
      FG(L,1)=FG(L,1)/100.00
      FG(L,2)=1.000-FG(L,1)
1    UC                                     )3.01F,X03,3.01F( TAMROF 001
C                                     SULF  C
100  FORMAT (10X,F10.3,20X,F10.0)         )0.01F,X01,3.01F,X02( TAMROF 001
C  IN
      READ (2,101) P
101  FORMAT (6F13.9)
      CALL LINK (PRFW1)
      END

```



```

// FOR PRFW1          CIRCUIT 57 AND 37 SIMULATIONS
  EXTERNAL PRFW2
C
C** PROGRAM TO PERFORM ACTUAL GAMMA FUNCTION MODEL PREDICTION OF TEST CIRCUIT
C OPERATION.  SEE PAGES 121-127
C** SUBROUTINE IAIB GENERATES CIRCUIT CONFIGURATION  S/R KROPH GENERATES
C QUADRATURE ROOTS AND WEIGHTS FOR INTEGRATION  S/R MINV5 PERFORMS INVERSION
C OF THE CONFIGURATION MATRIX
C** HANDLES A MAXIMUM OF 16 FEED CONDITIONS TO THE SAME CIRCUIT
C
C** X - CONTAINS THE RATIO OF CONCENTRATE VOLUME FLOW TO TAILS VOLUME FLOW FOR
C CALCULATION OF CELL MEAN RETENTION TIMES
C** RTZ,W - QUADRATURE ROOTS AND WEIGHTS FOR INTEGRATION
C** A,B - RATE DISTRIBUTION PARAMETERS
C** FG - SPECIES FRACTIONS IN FEED TO THE CIRCUIT
C** IA,IB - TAILS AND CONCENTRATE MATRICES RESPECTIVELY
C** RTG - PRODUCT OF THE TRANSPOSE OF THE R MATRIX AND THE G MATRIX.  SEE
C PAGE 135
C** C - CONFIGURATION MATRIX OR ITS INVERSE
C** P - RATE DISTRIBUTION PARAMETERS AS MEAN, VARIANCE AND NONFLOATABLE FRACTION
C** FV - FRESH FEED VOLUME FLOWRATES (L/MIN)
C** V - EFFECTIVE VOLUMES (LITERS) OF FLOTATION UNITS
C** TAU - MEAN RETENTION TIME IN A UNIT
C** TN - MASS FLOW OF SPECIES IN FINAL CONCENTRATE (ASSUMING 100 UNITS OF FEED)
C** GA - SPECIES GRADE IN FINAL CONCENTRATE
C** R - SPECIES RECOVERY IN THE TEST CIRCUIT AS PREDICTED
C
  DIMENSION X(40),RTZ(15),W(15),A(2),B(2),FG(16,2)
  DIMENSION IA(40,40),IB(40,40),RTG(40),C(40,40),P(3,2),FV(16),V(40)
  ,TAU(40)
  DIMENSION TN(16,2),GA(16),R(16,2)
  COMMON/INSKEL/MMM,A,B,FG,TN,GA,R,P,FV,TAU,RTZ,W,X
  COMMON IA,IB
  DATA V /20*2488.,20*622./
  DATA NXX /15/
  DO 3000 K=1,40
  DO 3000 KK=1,40
    IB(K,KK)=0
  3000 IA(K,KK)=0
  DO 8 M=1,2

```



```

      A(M)=P(1,M)**2./P(2,M)-1.00
      B(M)=P(1,M)/P(2,M)
      D1=RCGAM(A(M)+1.00)
      CALL KROPH (A(M),D1,RTZ,W,NXX)
      CALL IAIB (IA,IB,X)
      DO 4 L=1,MMM
      DO 50 I=1,40
      DO 50 J=1,40
50    C(I,J)=IA(I,J)+IB(I,J)*X(J)
      CALL MINV5 (C,40)
      DO 51 I=1,40
51    TAU(I)=V(I)/(-C(I,1)*FV(L))
      R(L,M)=0.
      DO 3 K=1,NXX
      DO 1 I=1,40
      RTG(I)=0.
      DO 2 J=1,40
2    RTG(I)=RTG(I)-IB(J,I)
1    RTG(I)=RTG(I)*RTZ(K)*TAU(I) /B(M)
      DO 6 I=1,40
      DO 6 J=1,40
6    C(I,J)=IA(I,J)+IB(I,J)*RTZ(K)*TAU(J)/B(M)
      CALL MINV5 (C,40)
      SUM=0.
      DO 7 I=1,40
7    SUM=SUM-C(I,1)*RTG(I)
      R(L,M)=R(L,M)+SUM*W(K)/D1
3    CONTINUE
      R(L,M)=(1.00-P(3,M))*R(L,M)
      TN(L,M)=FG(L,M)*100.*R(L,M)
4    CONTINUE
8    CONTINUE
      CALL LINK (PRFW2)
      END

```

```

// FOR PRFW2          OUTPUT FOR PRFW1
*IOCS(1443 PRINTER)
C
C** PROGRAM TO OUTPUT RESULTS FROM PRFW1
C** ZZ - RECOVERY OF VALUABLE COMPONENT
C** ZZZ - RECOVERY OF REMAINING MATERIAL (NONVALUABLE)
C** SS - RATIO OF RECOVERIES - VALUED/NONVALUED
C** Z11 - TAILS GRADE IN VALUES
C
  DIMENSION IA(40,40),IB(40,40),A(2),B(2),FG(16,2),TN(16,2),GA(16),R
    (16,2),P(3,2),FV(16),TAU(40),RTZ(15),W(15),X(40)
  COMMON/INSKEL/MMM,A,B,FG,TN,GA,R,P,FV,TAU,RTZ,W,X
  COMMON IA,IB
  DO 9 L=1,MMM
    GA(L)=TN(L,1)/(TN(L,1)+TN(L,2))*100.
    SS=R(L,1)/R(L,2)
    ZZ=R(L,1)
    ZZZ=R(L,2)
    Z11=(FG(L,1)*100.-TN(L,1))*100./(FG(L,2)*100.-TN(L,2))
  9  WRITE (3,101) ZZ,ZZZ,TN(L,1),TN(L,2),GA(L),SS,FV(L),Z11
101  FORMAT (8E12.4)
  WRITE (3,103) ((IA(I,J),J=1,40),I=1,40),((IB(I,J),J=1,40),I=1,40)
103  FORMAT (1H1,' TAILS MATRIX A',//40(40I2/),1H1,' CONCENTRATE MATRIX
    B',//40(1X,40I2/))
  CALL EXIT
  END

```



```
// FOR IAIB                                S/R FOR CIRCUIT 36
      SUBROUTINE IAIB (IA,IB,X)
C
C** USED TO GENERATE TAILS AND CONCENTRATE MATRICES FOR A ROUGHER,CLEANER AND
C   CLEANER MIDDLEING CIRCUIT.  USED BY PRFW1.
C** SEE PAGES 121-123
C
      DIMENSION IA(40,40),IB(40,40),X(40)
      DO 1 I=1,39
      X(I)=.011
      K=I+1
      IA(K,I)=1
      IA(I,I)=-1
1     IB(I,I)=-1
      IA(40,40)=-1
      IB(40,40)=-1
      IA(21,20)=0
      IA(1,40)=1
      DO 2 I=21,40
2     X(I)=.025
      DO 3 I=1,20
3     IB(21,I)=1
      DO 4 I=27,40
4     IB(21,I)=1
      RETURN
      END
```



```
// FOR          CIRCUIT 48   1ST RD 4 CL  
SUBROUTINE IAIB (IA,IB,X)
```

```
C  
C** USED TO GENERATE TAILS AND CONCENTRATE MATRICES FOR A ROUGHER, SCAVENGER,  
C CLEANER, CLEANER MIDDINGS CIRCUIT WHICH WAS TESTED. USED BY PRFW1.  
C** SEE PAGES 121-123
```

```
C  
  DIMENSION IA(40,40),IB(40,40),X(40)  
  DO 1 I=1,39  
    K=I+1  
    IA(K,I)=1  
    IA(I,I)=-1  
    IB(I,I)=-1  
1   X(I)=.011  
    IA(40,40)=-1  
    IB(40,40)=-1  
    IA(21,20)=0  
    IA(1,40)=1  
  DO 4 I=21,40  
4   X(I)=.025  
  DO 2 I=2,20  
2   IB(21,I)=1  
  DO 3 I=25,40  
3   IB(21,I)=1  
  RETURN  
  END
```

```
// FOR          CIRCUIT 53    NO CL MIDS
SUBROUTINE IAIB (IA,IB,X)
C
C** GENERATES TAILS AND CONCENTRATE MATRICES FOR A ROUGHER-CLEANER CIRCUIT
C WHICH WAS TESTED.  USED BY PRFW1
C** SEE PAGES 121-123
C
  DIMENSION IA(40,40),IB(40,40),X(40)
  DO 1 I=1,39
    K=I+1
    IA(K,I)=1
    IA(I,I)=-1
    IB(I,I)=-1
1  X(I)=.011
    IA(40,40)=-1
    IB(40,40)=-1
    IA(21,20)=0
    IA(1,24)=1
  DO 2 I=21,24
2  X(I)=.025
  DO 3 I=1,20
3  IB(21,I)=1
  RETURN
END
```



```
// FOR          CIRCUIT 61   RD CL RECL
SUBROUTINE IAIB (IA,IB,X)
C
C** GENERATES TAILS AND CONCENTRATE CONFIGURATION MATRICES FOR A ROUGHER,
C CLEANER AND RECLEANER CIRCUIT WHICH WAS TESTED.  USED BY PRFWL
C** SEE PAGES 121-123
C
  DIMENSION IA(40,40),IB(40,40),X(40)
  DO 1 I=1,39
    K=I+1
    IA(K,I)=1
    IA(I,I)=-1
    IB(I,I)=-1
1  X(I)=.011
    IA(40,40)=-1
    IB(40,40)=-1
    IA(21,20)=0
    IA(1,40)=1
  DO 4 I=21,40
4  X(I)=.025
    DO 2 I=1,20
2  IB(25,I)=1
    DO 3 I=27,40
3  IB(21,I)=1
  RETURN
END
```



//	XEQ	PRFWO	FX				
.4769		.466		0.	.002638	.0002154	0.
.065	.38		.164	1527.	3438.		
.077	.61		.264	1474.	3485.		
.071	.45		.182	1505.	3458.		
.068	.42		.18	1503.	3590.		
.075	.44		.195	1563.	3455.		
.063	.40		.185	1402.	3197.		
.073	.43		.166	1462.	3314.		
.069	.42		.168	1482.	3381.		
.078	.48		.187	1438.	3288.		
.096	.41		.174	1455.	3325.		
.074	.46		.214	1476.	3413.		
.076	.56		.187	1339.	3064.		
.074	.43		.135	1458.	3333.		
.084	.46		.198	1414.	3288.		
.093	.45		.219	1394.	3195.		
.098	.52		.228	1395.	3245.		

```

// FOR PRF22                CIRCUIT SIMULATION FOR GENERAL 25 UNITS
C
C** SIMULATION OF VARIOUS GENERAL FLOTATION CIRCUITS BY THE GAMMA FUNCTION FIRST
C ORDER MODEL.
C** THE METHOD USED IS OUTLINED IN PAGES 133-140
C** DD - CONTAINS TYPICAL PARAMETERS MEAN, VARIANCE AND NONFLOATING FRACTION FOR
C A SPECIES. I.E. RATE DISTRIBUTION PARAMETERS.
C** BY SUITABLE ARRANGEMENT OF THE OVERTURNED CARDS THE PROGRAM MAY BE USED TO
C EVALUATE A CIRCUIT AT FIVE TONNAGE RATES OR VARIATIONS OF THE SAME CIRCUIT
C AT A FIXED TONNAGE, I.E. A ROUGHER-SCAVENGER CIRCUIT WITH 1,2,3,4, AND 5
C CELLS AS FINAL CONCENTRATES.
C*****
C** PROGRAM IS NOW SETUP SO AS TO PERFORM AN EVALUATION OF SEVERAL CIRCUIT
C VARIATIONS AT A FIXED TONNAGE THROUGHPUT.
C*****
C** X - CONTAINS THE RATIO OF CONCENTRATE VOLUME FLOW TO TAILS VOLUME FLOW FOR
C EACH FLOTATION UNIT.
C** RTZ,W - CONTAINS THE ROOTS AND WEIGHTS FOR THE QUADRATURE FORMULA.
C** A,B - CONTAINS SPECIES DISTRIBUTION PARAMETERS IN THE FORM  $A = \text{MEAN}^2 / \text{VARIANCE}$ 
C  $E-1$ .  $B = \text{MEAN} / \text{VARIANCE}$ 
C** GA - CONTAINS THE CIRCUIT FINAL CONCENTRATE GRADES
C** R - RECOVERY OF THE TWO SPECIES
C** P - ALSO CONTAINS RATE CONSTANT DISTRIBUTION PARAMETERS
C** CONTAINS FEED GRADE OF THE SPECIES
C** IA,IB - TAILS AND CONCENTRATE MATRICES RESPECTIVELY
C** RTG - CONTAINS THE PRODUCT OF THE R MATRIX TRANSPOSED AND THE G MATRIX. SEE
C PAGE 135
C** C - CONTAINS THE CIRCUIT CONFIGURATION MATRIX OR ITS INVERSE
C** V,TAU - EFFECTIVE VOLUME OF EACH FLOTATION UNIT AND MEAN RETENTION TIME AS
C CALCULATED FOR EACH UNIT.
C** - DUPLICATE P ARRAY
C
      EXTERNAL PRF23
      DIMENSION X(25),RTZ(15),W(15),A(2),B(2),GA(5),R(5,2),P(3,2),FG(2)
      DIMENSION IA(25,25),IB(25,25),RTG(25),C(25,25),FD(5),V(25),TAU(25)
      DIMENSION DD(3,2)
      COMMON MMM,X,RTZ,W,A,B,FD,GA,R,P,TAU,RTG,V,FG
      COMMON IA,IB,NO
      DATA NXX /15/
C      UC

```



```

C      UC      / .0,4512000.,.836200.,.0,664.,9674./ DD ATAD
      DATA DD/.08643,.01329,0.,.002612,.0002175,0./      NI      C
      FG(1)=.002      NI      C
      NO=25
      MMM=5
      MMM=1      C
      FG(2)=1.0000-FG(1)

C      .01-I*.01+.08=)I(DF      18
C      5,1=I 18 OD

      DO 82 I=1,25
82      V(I)=2488.
      DO 83 I=1,3
      DO 83 J=1,2
83      P(I,J)=DD(I,J)
      FD(1)=106.      C
      DO 8 L=1,2
      L11=L*5+14
      DO 3000 K=1,25
      DO 3000 KK=1,25
      IB(K,KK)=0
3000      IA(K,KK)=0

C
C** CALL TO THE SUBROUTINE TO GENERATE CIRCUIT CONCENTRATE (IB) AND TAILS (IA)
C      MATRICES.
C
      CALL IAIB (IA,IB,X,L11)
      DO 8 M=1,2
      A(M)=P(1,M)**2./P(2,M)-1.00
      B(M)=P(1,M)/P(2,M)
      D1=RCGAM(A(M)+1.00)

C
C** CALL TO THE SUBROUTINE TO GENERATE THE ROOTS AND WEIGHTS NECESSARY FOR A
C      GAUSS-LAGUERRE QUADRATURE INTEGRATION FORMULA ON PAGE 140
C
      CALL KROPH (A(M),D1,RTZ,W,NXX)

C      MMM,1=L 4 OD

      DO 50 I=1,25
      DO 50 J=1,25
50      C(I,J)=IA(I,J)+IB(I,J)*X(J)
      CALL MINV5 (C,25)

```



```

      DO 51 I=1,NO
C          )766666.61*525./)L(DF*)1,I(C-(/)I(V=)I(UAT      15
C
C** CALCULATION OF EACH FLOTATION UNITS MEAN RETENTION TIME.
C
51  TAU(I)=V(I)/(-C(I,1)*FD(1)/.525*16.66667)
    R(L,M)=0.
    DO 3 K=1,NXX
    DO 1 I=1,NO
    RTG(I)=0.
    DO 2 J=1,NO
2   RTG(I)=RTG(I)-IB(J,I)
1   RTG(I)=RTG(I)*RTZ(K)*TAU(I) /B(M)
    DO 6 I=1,NO
    DO 6 J=1,NO
6   C(I,J)=IA(I,J)+IB(I,J)*RTZ(K)*TAU(J)/B(M)
    CALL MINV5 (C,25)
    SUM=0.
    DO 7 I=1,NO
7   SUM=SUM-C(I,1)*RTG(I)
    R(L,M)=R(L,M)+SUM*W(K)/D1
3   CONTINUE
    R(L,M)=(1.00-P(3,M))*R(L,M)
C
C          )2,L(R*)L(DF*)2(GF=BT
          )1,L(R*)L(DF*)1(GF=AT
C
    TA=FG(1)*FD(1)*R(L,1)
    TB=FG(2)*FD(1)*R(L,2)
    GA(L)=TA/(TA+TB)*100.
4   CONTINUE
8   CONTINUE
    CALL LINK (PRF23)
    END

```

```

// FOR PRF23          OUTPUT FOR PRF22
*IOCS(1443 PRINTER)
C
C** OUTPUT PROGRAM FOR GENERAL FLOTATION CIRCUIT SIMULATION PROGRAM PRF22.
C** ARRAY DESIGNATIONS THE SAME AS FOR PRF22.
C
  DIMENSION X(25),RTZ(15),W(15),A(2),B(2),FD(5),GA(5),R(5,2),P(3,2),
  TAU(25),RTG(25),V(25),FG(2)
  DIMENSION IA(25,25),IB(25,25)
  COMMON MMM,X,RTZ,W,A,B,FD,GA,R,P,TAU,RTG,V,FG
  COMMON IA,IB,NO
  MMM=5
  WRITE (3,100) P,A,B,FG,TAU
100  FORMAT (1H1,' PARAMETERS  '/1X,10E11.4/' FEED GRADE '/2E12.2/' RES
      TIME '/3(10E11.4/)/)
  WRITE (3,101) (FD(L),R(L,1),R(L,2),GA(L),L=1,MMM)
101  FORMAT (4E16.6)
  WRITE (3,103) ((IA(I,J),J=1,NO),I=1,NO),((IB(I,J),J=1,NO),I=1,NO)
103  FORMAT  ((' TAILS MATRIX A'//25(1X,25I2/))' CONCENTRATE MATRIX B
      '/25(1X,25I2/))
  CALL EXIT
  END

```



```

// FOR IAIB
SUBROUTINE IAIB (IA,IB,X,N2)
C
C** IN THIS GENERAL CIRCUIT SIMULATION SUBROUTINE A MAXIMUM NUMBER OF TWENTY-FIVE
C FLOTATION UNITS IS AVAILABLE, EACH CIRCUIT OR CIRCUIT VARIATION USES ALL OF
C THE CELLS.
C
C     DIMENSION IA(25,25),IB(25,25),X(25)
C     DO 1 I=1,25
C       IA(I,I)=-1
C       IB(I,I)=-1
C     1   X(I)=.011
C
C*****
C** SUBROUTINE TO GENERATE CONCENTRATE (IB) AND TAILINGS (IA) MATRICES FOR A
C GENERAL ROUGHER SCAVENGER CIRCUIT. SEE PAGES 116-119
C** X - CONTAINS THE RATIO OF CONCENTRATE VOLUME FLOW TO TAILS VOLUME FLOW FOR
C EACH CELL. ESTIMATED TO CALCULATE CIRCUIT RESIDENCE TIMES.
C** N2 - NUMBER OF ROUGHER CELLS, I.E. FINAL CONCENTRATE CELLS.
C
C     NN=N2
C     NX=NN+1
C     DO 2 I=1,24
C       K=I+1
C     2   IA(K,I)=1
C     DO 3 I=NX,25
C     3   IB(1,I)=1
C
C*****
C** CARDS FOR ROUGHER-CLEANER-RECLEANER CIRCUIT.
C** N2 - NUMBER OF RECLEANER CELLS NUMBER OF CLEANERS IS FIXED AT FOUR.
C
C     NN=25-N2-4
C     NX=NN-1
C     NO=N2
C     NY=NN+NO+1
C     NZ=NN+1
C     DO 2 I=1,NX
C       K=I+1
C     2   IA(K,I)=1

```

```

ROSCAV C
ROSCAV C
ROSCAV C
ROSCAV C
ROSCAV C
ROSCAV C
ROSCAV C
RCREC C
RCREC C
RCREC C
RCREC C
RCREC C
RCREC C
RCREC C

```



```

      DO 3 I=1,NN
3      IB(NY,I)=1
      DO 4 I=NY,25
4      IB(NZ,I)=1
      DO 5 I=NZ,24
      K=I+1
      X(I)=.025
      IA(K,I)=1
5      IA(1,25)=1
      X(25)=.025
C
C*****
C
C** CARDS FOR ROUGHER-CLEANER-CLEANER MIDDLEINGS CIRCUIT.
C** N2 - NUMBER OF CLEANER CELLS  NUMBER OF CLEANER MIDDLEINGS CELLS IS FIXED AT
C    FOUR.
      NN=25-N2-4
      NX=NN-1
      NZ=NN+1
      NO=N2
      NY=NO+NZ
      DO 2 I=1,NX
      K=I+1
2      IA(K,I)=1
      DO 3 I=1,NN
3      IB(NZ,I)=1
      DO 4 I=NZ,24
      K=I+1
      X(I)=.025
4      IA(K,I)=1
      IA(1,25)=1
      X(25)=.025
      DO 5 I=NY,25
5      IB(NZ,I)=1
C
C*****
C** CARDS FOR ROUGHER-CLEANER CIRCUIT.
C** N2 - NUMBER OF CLEANER CELLS.
      NN=25-N2

```

RCREC C  
RCREC C  
RCREC C  
RCREC C  
RCREC C  
RCREC C  
RCREC C  
RCREC C

RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C  
RCM C

ROCL C

```

      NX=NN-1
      NZ=NN+1
      DO 2 I=1,NX
        K=I+1
2       IA(K,I)=1
        DO 3 I=1,NN
3       IB(NZ,I)=1
        IA(1,25)=1
        DO 4 I=NZ,25
4       X(I)=.025
        IF(NZ-1) 6,6,7
7       DO 5 I=NZ,24
        K=I+1
5       IA(K,I)=1
6       CONTINUE
      RETURN
      END

```



```

// FOR RESTT                                RESIDENCE TIME VOL ESTIMATION
*IOCS(TYPEWRITER,KEYBOARD)
C
C** PROGRAM TO BEST ESTIMATE THE EFFECTIVE VOLUME OF A KNOWN NUMBER OF STIRRED
C  UNITS WHICH ARE ASSUMED TO BE PERFECTLY MIXED
C** Y - OBSERVED TRACER CONCENTRATIONS AT TAILS EXIT
C** T - CORRESPONDING TIME OF OBSERVATION WITH THE INPUT PULSE AS TIME ZERO
C** V - INITIAL EFFECTIVE VOLUME ESTIMATE FOR ALL UNITS. ON EXIT V IS THE BEST
C  ESTIMATED EFFECTIVE VOLUME
C** P - PREDICTED TRACER CONCENTRATION AT TIME T
C** S - SUM OF SQUARED DEVIATIONS
C
C** COULD BE SIMPLIFIED BY COMBINATION WITH USEIN
C
      DIMENSION T(25),Y(25),P(25)
      CALL LARRY (25,NV,Y)
      CALL LARRY (25,NV,T)
      N=25-NV
      ST=100.
      W=10.**10.
      V=37000.
      V=17000.
      V=10500.
3     V=V+ST
      CALL SSQS (T,Y,P,N,V,S)
      IF(S-W) 1,2,2
1     W=S
      GO TO 3
2     V=V-2.*ST
      ST=ST/10.
      IF(ST-.2) 99,99,4
4     CALL SSQS (T,Y,P,N,V,S)
      W=S
      GO TO 3
99    CALL SSQS (T,Y,P,N,V,S)
      WRITE (3,100) (T(I),Y(I),P(I),I=1,25),V,S
100   FORMAT (25(3E14,4/)/2E12.4)
      CALL EXIT
      END

```



```
// FOR USEIN          USE OF INTEGRATION QTFG
*IDCS(TYPEWRITER,KEYBOARD)
C
C** MAINLINE DATA INPUT PROGRAM FOR INTEGRATION BY TRAPAZOIDAL RULE
C** USED TO OBTAIN AREAS UNDER RESIDENCE TIME DISTRIBUTIONS
C** SEE PAGES 37-41
C
  DIMENSION X(25),Y(25),Z(25)
1  CONTINUE
  CALL LARRY (25,NV,X)
  CALL LARRY (25,NV,Y)
  NDIM=25-NV
  CALL QTFG (X,Y,Z,NDIM)
  WRITE (1,100) Z
100 FORMAT (5E12.4)
  GO TO 1
99  CALL EXIT
  END
```

```

// FOR BATPL
*IOCS(1443 PRINTER)
*IOCS(TYPEWRITER,KEYBOARD)
C
C** PROGRAM TO CALCULATE NORMALIZED BATCH TEST CONCENTRATION VERSUS TIME CURVES
C FROM CONCENTRATE AND TAIL ASSAYS AND WEIGHTS
C** IF SULFUR ASSAYS ARE ENTERED A NON-SULFIDE CONCENTRATION CURVE IS ALSO
C OUTPUT
C** WT - CONTAINS CONCENTRATES AND FINAL TAIL WEIGHTS
C** A - CONTAINS CONCENTRATE AND FINAL TAIL ASSAYS
C** CONTAINS NORMALIZED CONCENTRATION VALUES
C
      DIMENSION IALFA(40)
      DIMENSION A(15),WT(15),W(15),S(15)
C** CALCULATES MINERAL AND GANGUE CURVES
100  FORMAT (I3)
101  FORMAT (80A2)
102  FORMAT (8F10.4)
7    READ (4,100) N
      IF(N) 99,99,8
8    READ (4,101) IALFA
C
C** FREE FORMAT TYPEWRITER INPUT TO ARRAY WT
C
      CALL RINP (IALFA,1,78,15,NVRD,WT)
      READ (4,101) IALFA
      CALL RINP (IALFA,1,78,15,NVRD,A)
5    SUMS =0
      DO 1 I=1,N
        W(I)=A(I)*WT(I)/100.
1    SUMS = SUMS + W(I)
      K=N-1
      DO 2 I=1,K
        X=0.
        DO 3 J=1,I
3        X=X+W(J)/SUMS
2        S(I)=1.0-X
      WRITE (1,102) (S(I),I=1,N)
      IF (A(1)-50.)6,7,7
6    DO 4 I=1,N

```



```
4  A(I)=100.-A(I)
   GO TO 5
99 CALL EXIT
   END
```



```

// FOR PYRRH          CALCS PYRRHOTITE CURVE FROM CU,NI,S
*IOCS(1443 PRINTER)
*IOCS(CARD)
C
C** PROGRAM TO GIVE PYRRHOTITE CURVE FROM CU,NI,AND S ASSAYS.  PROGRAM ASSUMES
C  CHALCOPYRITE TO BE CUFE2, PENTLANDITE TO BE (FE,NI)9S8, AND PYRRHOTITE TO BE
C  FE7S8.  34.6 CU, 30.4 FE, 35.0 S IN CHALCOPYRITE $ 39.1 FE, 41.2 NI, 19.7 S
C  IN PENTLANDITE $ 60.4 FE, 39.6 S IN PYRRHOTITE.
C
      DIMENSION A(13,5),S(5),WT(13),T(20)
97    READ (2,100) T
      DO 10 L=1,5
10     S(L)=0.
      WRITE (3,103)
      WRITE (3,100) T
      WRITE (3,103)
100    FORMAT (20A4)
101    FORMAT (5F10.4)
C
C** READ IN KNOWN WEIGHTS IN SAMPLE AND KNOWN SULFUR, COPPER, AND NICKEL ASSAYS
C
      CALL RDCRD (13,NV,WT)
      CALL RDCRD (13,NV,A(1,1))
      CALL RDCRD (13,NV,A(1,2))
      CALL RDCRD (13,NV,A(1,3))
      N=13-NV
      DO 1 I=1,N
      X=WT(I)*A(I,1)
      XX=35./34.6*WT(I)*A(I,2)
      XXX=19.7/41.2*WT(I)*A(I,3)
1     A(I,4)=(X-XX-XXX)/WT(I)*60.4/39.6
C
C** OUTPUT CALCULATED IRON AS PYRRHOTITE ASSAYS
C
      WRITE (3,101) (WT(I),A(I,1),A(I,2),A(I,3),A(I,4),I=1,N)
      IF(A(N,4)) 40,41,41
40     A(N,4)=0.
41     CONTINUE
      DO 2 J=1,4
      DO 2 I=1,N

```

```
2      S(J)=S(J)+A(I,J)*WT(I)
      DO 6 I=1,N
6      S(5)=S(5)+WT(I)*(100.-A(I,1))
      DO 3 I=1,N
      A(I,5)=(100.-A(I,1))*WT(I)/S(5)
      DO 3 J=1,4
3      A(I,J)=A(I,J)*WT(I)/S(J)
      L=N-1
      DO 4 J=1,5
      DO 4 I=1,L
      K=N-I
4      A(K,J)=A(K+1,J)+A(K,J)
      WRITE (3,103)
103    FORMAT (//)
C
C** OUTPUT NORMALIZED CONCENTRATION CURVES FOR SULFUR, NONSULFIDE, COPPER,
C    NICKEL AND PYRRHOTITE SPECIES
C
      WRITE (3,102) ((A(K,J),J=1,5),K=1,N)
102    FORMAT (5F10.4)
      GO TO 97
98    CALL EXIT
      END
```



```
// FOR QTFG                                TRAPIZOIDAL RULE INTEGRATION
      SUBROUTINE QTFG (X,Y,Z,NDIM)
C
C** IBM SUPPLIED SUBROUTINE FOR INTEGRATION OF TABULATED FUNCTIONS * USED BY
C  USEIN.
C** X - CONTAINS ABCISSA DATA
C** Y - CONTAINS ORDINATE DATA
C** Z - CONTAINS STEPWISE AREA UNDER CURVE ** Z(NDIM) CONTAINS TOTAL AREA
C
      DIMENSION X(1),Y(1),Z(1)
      SUM2=0.
      IF (NDIM-1) 4,3,1
1     DO 2 I=2,NDIM
        SUM1=SUM2
        SUM2=SUM2+.5000*(X(I)-X(I-1))*(Y(I)+Y(I-1))
2     Z(I-1)=SUM1
3     Z(NDIM)=SUM2
4     RETURN
      END
```



```
// FOR SSQS                                WITH RESTT
      SUBROUTINE SSQS (T,Y,P,N,V,S)
C
C** SUBROUTINE FOR RESTT
C** C - NUMBER OF STIRRED UNITS
C** SV - FLOWRATE THROUGH STIRRED UNITS IN UNITS OF T AND V
C
      DIMENSION T(1),Y(1),P(1)
      S=0.
      C=20.
      SV=870.
      DO 1 I=1,N
      E=C*SV/(V*RFACT(C-1.))*(C*SV*T(I)/V)**(C-1.)*EXP(-C*SV*T(I)/V)
      P(I)=E*27.64
C
C** THE ABOVE CONSTANT (27.64) IS THE AREA UNDER THE OBSERVED CONCENTRATION
C    VERSUS TIME CURVE IN UNITS OF Y AND T. CAN BE CALCULATED FROM PROGRAM USEIN
C
      1  S=S+(Y(I)-P(I))**2.
      RETURN
      END
```

```
// FOR BREAD                                READ IN BATCH DATA
      SUBROUTINE BREAD (T,Y,IS,ID,NDATA)
C
C** SUBROUTINE FOR INPUT OF CONCENTRATION CURVE DATA * FIRST SECTION OF THESIS
C  PART TWO
C** T - AN ARRAY OF LENGTH NDATA CONTAINING TIMES
C** Y - AN ARRAY OF LENGTH NDATA CONTAINING NORMALIZED CONCENTRATIONS OF A
C  SPECIES CORRESPONDING TO ARRAY T
C** IS - AN ALPHANUMERIC ARRAY OF 60 CHARACTERS IDENTIFYING THE TEST
C** ID - AN INTEGER ARRAY OF SEVEN WORDS CONTAINING THE TEST CODE
C** NDATA - NUMBER OF PAIRS OF CONCENTRATIONS AND TIMES
C
      DIMENSION T(1),Y(1),ID(7),IS(30)
      READ (2,201) IS,ID,NDATA
201  FORMAT (30A2,6X,3I1,I2,I4,I1,I4/2X,I3)
      READ (2,202) (Y(I),I=1,NDATA)
202  FORMAT (20F4.3)
      READ (2,203) (T(I),I=1,NDATA)
203  FORMAT (20F4.2)
      RETURN
      END
```



```
// FOR MINV5          MATRIX INVERSION
      SUBROUTINE MINV5 (C,M)
C
C** MATRIX INVERSION SUBROUTINE FOR THE GAMMA FUNCTION MODEL CIRCUIT SIMULATIONS
C  AND ANYFT REGRESSIONS
C  ALSO USED BY GCONT
C
      DIMENSION C(40,40)
C  DIRECT MATRIX INVERSION LIB. PROG. 5.0.CE003
      DO 5014 I = 1,M
        W = C(I,I)
        C(I,I) = 1.0
        DO 5011 J = 1,M
5011  C(I,J) = C(I,J) / W
        DO 5014 K = 1,M
          IF (K - I) 5012,5014,5012
5012  W = C(K,I)
          C(K,I) = 0.
          DO 5013 J = 1,M
5013  C(K,J) = C(K,J) - W * C(I,J)
5014  CONTINUE
        RETURN
      END
```



```
// FOR RCGAM          IBM GAMMA FUNCTION
      FUNCTION RCGAM (XX)
```

```
C
C** IBM SUPPLIED FUNCTION SUBPROGRAM FOR THE EVALUATION OF THE GAMMA FUNCTION
```

```
C
      IF (XX-31.) 6,6,4                                MACHSPEC
4      RCGAM=10.**10.
      RETURN
6      X=XX
      GX=1.0
      IF (X-2.0) 50,50,15
10     IF (X-2.0) 110,110,15
15     X=X-1.0
      GX=GX*X
      GO TO 10
50     IF (X-1.0) 60,120,110
60     IF (X-1.0E-6) 62,62,80
62     Y=FLOAT(IFIX(X))-X
      IF (ABS(Y)-1.0E-6) 130,130,64
64     IF (1.0-Y-1.0E-6) 130,130,70
70     IF (X-1.0) 80,80,110
80     GX=GX/X
      X=X+1.0
      GO TO 70
110    Y=X-1.0
      GY=1.+Y*(-0.5771017+Y*(0.9858540+Y*(-0.8764218+Y*(.8328212+Y*(-.56
      84729+Y*(.2548205+Y*(-.05149930))))))
      GX=GX*GY
120    CONTINUE
130    CONTINUE
      RCGAM=GX
      RETURN
      END
```

```

// FOR KROPH          ROOTS/WEIGHTS GEN LAQUERRE QUADRATURE
      SUBROUTINE KROPH (ALF,CC,X,A,NN)
C
C** GENERATES ROOTS AND WEIGHTS FOR GENERALIZED GAUSS-LAGUERRE QUADRATURE. THE
C  MAXIMUM DEGREE OF POLYNOMIAL POSSIBLE IS MACHINE SPECIFIC AS WELL AS PARTLY
C  DEPENDENT ON THE VALUE OF ALF
C** SOLVES FOR ROOTS AND WEIGHTS IN THE GENERAL INTEGRATION PROBLEM PRESENTED ON
C  PAGE 140
C** COURTESY OF DR. I. J. KROPHOLLER, LOUGHBUROUGH UNIVERSITY
C** ALF - VALUE OF THE POWER OF X IN THE GENERAL INTEGRAL FORM
C** CC - GAMMA FUNCTION OF ALF+1.
C** X - ROOTS OF THE QUADRATURE FORMULA
C** A - WEIGHTS OF THE QUADRATURE FORMULA
C** NN - DEGREE OF THE POLYNOMIAL REQUIRED
C
      DIMENSION X(15),A(15),B(15),C(15)
C
C          )52(C, )52(B, )52(A, )52(X NOISNEMID
C          )23(C, )23(B, )23(A, )23(X NOISNEMID
C
      EPS=1.E-9
      DO 1 I=2,NN
      XI=I
      B(I)=(ALF+2.*XI-1.)
      C(I)=(XI-1.)*(ALF+XI-1.)
      CSX=CC
      CALL LAGUR (NN,X,A,ALF,B,C,EPS,CSX,CSA,TSX,TSA)
      RETURN
      END

```

KRP C  
TEMP







```
7      CSA=CSA+A(I)  
      CONTINUE  
      RETURN  
      END
```

```

// FOR LROOT          S/R FOR QUADD
SUBROUTINE LROOT (X,NN,ALF,DPN,PN1,B,C,EPS)
C
C** SUBROUTINE TO CALCULATE ROOTS OF THE QUADRATURE POLYNOMIAL USED BY S/R
C KROPH.
C** COURTESY OF DR. I. J. KROPHOLLER, LOUGHBUROUGH UNIVERSITY
C
C      DIMENSION B(32),C(32)
C
C
C      )52(C, )52(B NOISNEMID
C      )51(C, )51(B NOISNEMID
C
C      ITER=0
1     ITER=ITER+1
      CALL LRECR (P,DP,PN1,X,NN,ALF,B,C)
      D=P/DP
      X=X-D
      IF (ABS(D/X)-EPS) 3,3,2
2     IF (ITER-20) 1,3,3
3     DPN=DP
      RETURN
      END

```

```

// FOR LRECR                      S/R FOR QUADD
      SUBROUTINE LRECR (PN,DPN,PN1,X,NN,ALF,B,C)
C
C** SUBROUTINE TO CALCULATE VALUE AND DERIVATIVE OF THE GENERALIZED LAGUERRE
C   POLYNOMIALS USING THE RECURSIVE RELATIONSHIP.  USED BY S/R KROPH
C** COURTESY OF DR. I. J. KROPHOLLER, LOUGHBUROUGH UNIVERSITY
C
      DIMENSION B(32),C(32)
C
C                                     )52(C, )52(B NOISNEMID
C                                     )51(C, )51(B NOISNEMID
C
      P1=1.
      P=X-ALF-1.
      DP1=0.
      DP=1.
      DO 1 J=2,NN
      Q=(X-B(J))*P-C(J)*P1
      DQ=(X-B(J))*DP+P-C(J)*DP1
      P1=P
      P=Q
      DP1=DP
1     DP=DQ
      PN=P
      DPN=DP
      PN1=P1
      RETURN
      END

```



```
// FOR RFACT
      FUNCTION RFACT (N)
C
C** ROUTINE USED TO CALCULATE FACTORIAL EXPRESSIONS
C
      IF (N-1) 4,4,3
4      RFACT=1.
      GO TO 99
3      RFACT=N
      K=N-1
      DO 1 I=1,K
1      RFACT=RFACT*FLOAT(N-I)
99     RETURN
      END
```

```

// FOR MODSQ
  SUBROUTINE MODSQ (A,B,S)
C
C** SUBROUTINE FOR USE WITH ITIT OR SEAR2
C
  COMMON Y(13),T(13),PRED(13),ISA(50),ID(7),IS(30),ILC(7),AA(2),BB(2
1),Q(13),NDATA,DB,DA,IWT(2),SG(13)
  S=0.
  W1=Y(NDATA)
  W2=.5*(1.-Y(2))
C
  PRED(1)=1.
  DO 1 I=2,NDATA
    Z=A+B+Q(I)
    X=4*A*Q(I)
    P=SQRT(Z**2.-X)
    E=(-Z+P)/2.
    F=(-Z-P)/2.
    Z=E-F
    ZEDD=T(I)-T(I-1)
    PRED(I)=(E*EXP(F*ZEDD)-F*EXP(E*ZEDD))/Z
    PRED(I)=PRED(I-1)*PRED(I)
1  CONTINUE
  DO 2 I=2,NDATA
    PRED(I)=(1.-IWT(1)*W1)*PRED(I)+IWT(1)*W1
2  S=S+(Y(I)-PRED(I))**2.
  RETURN
  END

```

)I(GS/B=B

